FINAL REPORT—ARSENIC REMOVAL DEMONSTRATION PROJECT MINE WASTE TECHNOLOGY PROGRAM ACTIVITY III, PROJECT 9

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Prepared for:

U.S. Environmental Protection Agency National Risk Management Research Laboratory Office of Research and Development Cincinnati, Ohio 45268 IAG ID No. DW89935117-01-0

and

U.S. Department of Energy Federal Energy Technology Center Pittsburgh, Pennsylvania 15236 Contract No. DE-AC22-96EW96405

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FINAL REPORT—ARSENIC REMOVAL DEMONSTRATION PROJECT

MINE WASTE TECHNOLOGY PROGRAM ACTIVITY III, PROJECT 9

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Foreword

Today, the mineral industries are developing and modifying technologies that will enable industries to operate more efficiently. If improperly dealt with, the waste generated by these industries can threaten public health and degrade the environment. The U.S. Environmental Protection Agency (EPA) is charged by the Congress of the United States with protecting the Nation's land, air, and water resources. Under a mandate of national environmental laws, the EPA strives to formulate and implement actions leading to a balance between human activities and the ability of natural systems to support and nurture life. These laws direct the EPA to perform research to define, measure the impacts, and search for solutions to environmental problems.

The National Risk Management Research Laboratory (NRMRL) of EPA is responsible for planning, implementing, and managing research, development, and demonstration programs to provide an authoritative, defensible engineering basis in support of the policies, programs, and regulations of the EPA with respect to drinking water, wastewater, pesticides, toxic substances, solid and hazardous wastes, and Superfund-related activities. The Federal Energy Technology Center (FETC) of the U.S. Department of Energy (DOE) has responsibilities similar to the NRMRL in that FETC is one of several DOE centers responsible for planning, implementing, and managing research and development programs. This document is a product of the research conducted by these two Federal organizations.

This document is the final report for EPA's Mine Waste Technology Program (MWTP) Activity III, Project 9, Arsenic Removal Demonstration Project. The MWTP is a program developed through an Interagency Agreement between EPA and DOE. MSE Technology Applications, Inc. manages the MWTP and is responsible for the field demonstration activities and preparing this document. The information generated under this program provides a vital communication link between the researcher and the user community.

One of the objectives of the MWTP is to identify the types of mining wastes impacting the nation and the technical issues that need to be addressed. Other objectives of this program are: 1) address these technical issues through application of treatment technologies, 2) determine the candidate technologies that will be tested and evaluated, and 3) determine the candidate waste form/sites where these evaluations will take place.

Executive Summary

This document is the final report for the U.S. Environmental Protection Agency's (EPA) Mine Waste Technology Program (MWTP) Activity III Project 9, Arsenic Removal Demonstration Project. The MWTP is a program developed through an Interagency Agreement (IAG) between EPA and the U.S. Department of Energy (DOE). MSE Technology Applications, Inc. (MSE) manages the MWTP and owns/operates the MSE Testing Facility in Butte, Montana, previously the DOE–Western Environmental Technology Office. MSE proposed and was granted funding for the Arsenic Removal Demonstration Project during the December 1996 IAG Management Committee Meeting.

Acidic, metal-bearing water draining from remote abandoned mines has been identified by the EPA as a significant environmental/health hazard in the Western United States. Many of these waters contain dissolved arsenic in the trivalent and pentavalent state. The arsenic problems in discharge streams are directly related to the EPA's Technical Issue *Mobile Toxic Constituents—Water*. The National Drinking Water Standard for arsenic is 50 parts per billion (ppb). The World Health Organization revised the guideline for arsenic in drinking water from 50 to 10 ppb in 1993.

The purpose of the Arsenic Removal Demonstration Project was to demonstrate alternative treatment technologies capable of removing arsenic from mineral industry effluents to below 50 ppb. Several technologies with potential application to treat arsenic problems were presented in the MWTP Activity I, Volume 5, *Issues Identification and Technology Prioritization Report—Arsenic*. Each technology was screened and prioritized on the basis of its potential to reduce arsenic levels in the mineral industry. Two innovative technologies were selected, Mineral-Like Precipitation and Alumina Adsorption with Microfiltration. Both technologies were demonstrated/evaluated by treating two of the same industrial effluents, industrial process water and arsenic-contaminated mine water. The Ferrihydrite Adsorption technology, EPA's Best Demonstrated Available Technology (BDAT) for removal of arsenic, was used for comparative purposes.

In January 1997, MSE prepared agreements with Montana Tech of the University of Montana (Montana Tech) and ZENON Environmental, Inc. (ZENON). These agreements were signed for the demonstration/evaluation of their Mineral-Like Precipitation and Alumina Adsorption with Microfiltration technologies respectively. Four Montana Tech employees and two ZENON employees, in collaboration with MSE staff, performed the pilot–scale demonstrations in July–September 1997. This report addresses the results of the pilot demonstration projects and the subsequent leachability testing of the arsenical residues produced during the demonstration.

Technology Demonstrations

Mineral-Like Precipitation

The concept of this process is to strip arsenic from solutions in such a manner so as to produce mineral-like precipitated salts that are stable for long-term storage in outdoor pond-type environments. This process was developed by the Principal Investigator, Dr. Larry Twidwell, and may be accomplished by precipitation from solutions containing arsenate and phosphate. The concept was to substitute arsenate into an apatite structure $[Ca_{10}(PO_4)_6(OH)_2]$ thereby forming a solid solution

compound $[Ca_{10}(As_xP_yO_4)_6(OH)_2]$ that would be thermodynamically stable in an outdoor storage environment.

Alumina Adsorption

Alumina adsorption technology uses aluminum oxide to adsorb arsenic onto its surface from arsenic—bearing solutions. The process is completed at a certain pH range. After absorption, reagents are added to the alumina to desorb the arsenic from the solid into a concentrated brine. The concentrated arsenic brine solution is then treated using an iron adsorption technology to remove and stabilize the arsenic. The activated alumina in the process is recycled following the desorption process where it is treated with a strong caustic solution of sodium hydroxide.

Ferrihydrite Adsorption

Ferrihydrite technology is an industrial technique commonly used for dissolved heavy metal removal and, as stated earlier, is EPA's BDAT for arsenic removal. For ferrihydrite adsorption to occur, the ferric iron (Fe^{+3}) must be present in the water to be treated. Dissolved arsenic is removed by a lime neutralization process in the presence of the ferric iron, which results in the formation of arsenic–bearing hydrous ferric oxide (ferrihydrite).

Waste Stream Description

Potential waste streams were identified and prioritized in the MWTP Activity I, Volume 5, Appendix A, Issues Identification and Prioritization for Arsenic. The arsenic removal demonstration was designed to demonstrate arsenic removal technologies that are capable of removing arsenic to below the federal discharge standards of 50 ppb. Three different waters were treated, e.g., ASARCO's lead smelter scrubber blowdown water [containing > 3 grams per liter (g/L) arsenic and many other associated metals], ASARCO's water treatment thickener overflow water [containing \sim 6 parts per million (ppm) arsenic], and TVX Mineral Hill Mine 1,300' Portal groundwater (containing \sim 500 ppb arsenic).

Demonstration Results

All three addressed technologies (ferrihydrite adsorption, alumina adsorption, and mineral-like precipitation) showed favorable results for arsenic removal using groundwater; however, using industrial process wastewater, only two of the technologies (Mineral-Like Precipitation and Ferrihydrite Adsorption) were capable of removing arsenic to below necessary discharge standards. The complex chemistry of the industrial wastewater had a profound effect on arsenic removal using alumina adsorption.

Mineral-Like Precipitation

Mineral-Like Precipitation removed significantly more of the arsenic in each of the demonstrations than the stated goal of the project [i.e., to lower the arsenic content in the effluent water to less than the drinking water standard for arsenic (< 50 ppb)]. In fact, the final arsenic content in the effluent waters was in most cases < 10 ppb. A summary of the results for each demonstration is presented in Table ES-1.

Table ES-1. Mineral-Like Precipitation Results

Creatons	Description	[As] Concentration	
System	Description	Inlet Water	Effluent Water
ASARCO Scrubber Blowdown Water	P/As \sim 5.5, 1,665 gallons treated	~ 3.3 g/L	7-9 Fg/L
ASARCO Scrubber Blowdown Water	P/As~ 12, 405 gallons treated	~ 3.3 g/L	6-9 Fg/L
ASARCO Thickener Overflow Water	P/As~ 10, 1,185 gallons treated	\sim 5.8 mg/L	6-15 Fg/L
ASARCO Thickener Overflow Water	P/As~ 100, 1,425 gallons treated	~ 5.8 mg/L	3-13 Fg/L
Mineral Hill Mine 1,300' Portal Groundwater	P/As~ 10, 1,185 gallons treated	420 Fg/L	6-7 Fg/L
Mineral Hill Mine 1,300' Portal Groundwater	P/As \sim 20, 3,915 gallons treated	450 Fg/L	4-7 Fg/L

Alumina Adsorption

Alumina adsorption technology was very successful in removing arsenic when treating TVX's Mineral Hill Mine 1,300' Portal groundwater (containing ~ 500 ppb arsenic). Treating ASARCO's lead smelter thickener overflow water (containing ~ 6.0 mg/L arsenic and many other associated metals) with this technology is ineffective. Other species (e.g., sulfate) competed and interfered with available alumina adsorption sites. A summary of the results for each demonstration is presented in Table ES-2.

Table ES-2. Alumina Adsorption Results

Crestons	[As] Con	Concentration	
System	Inlet Water	Effluent Water	
ASARCO Thickener Overflow Water (60 g/L Activated Alumina)	$\sim 6.0~g/L$	\sim 200 mg/L	
Mineral Hill Mine 1,300' Portal Groundwater	450 Fg/L	21 Fg/L	

Ferrihydrite Adsorption

Ferrihydrite adsorption technology was successful in treating both of the demonstration waters. Using an iron/arsenic mole ratio of 8 produced adequate results; however, the arsenic drinking water discharge standard of less than 50 ppb was never achieved treating the thickener overflow water. Increasing the iron to arsenic mole ratio to 10 when treating both the thickener overflow water and the Mineral Hill Mine 1,300' Portal groundwater lowered arsenic concentrations to less than discharge standards. A summary of the results for each demonstration is presented in Table ES-3.

Table ES-3. Ferrihydrite Adsorption Results

Crystom	[As] Co	oncentration
System	Inlet Water	Effluent Water
ASARCO Thickener Overflow Water (Iron to Arsenic Mole Ratio = 8)	6.0 mg/L	~ 100 Fg/L
ASARCO Thickener Overflow Water (Iron to Arsenic Mole Ratio = 10)	6.0 mg/L	~ 20 Fg/L
Mineral Hill Mine 1,300' Portal Groundwater	450 Fg/L	< 50 Fg/L

Economic Evaluation

One objective of this study was to perform a first–order cost estimate for the developed treatment flowsheets. Therefore, a "first–order" cost estimate was performed. The cost estimate presented here is not a detailed engineering cost analysis. It is a first–order cost estimate that should be within \pm 30%.

Table ES-4. Economic Evaluation for Selected Technologies Treating Groundwater with 500 ppb Arsenic at 300 gpm.

	Mineral-Like Precipitation	Alumina Adsorption	Ferrihydrite Adsorption
Capital	$$250,000 \pm 75,000$	$$396,000 \pm 118,8000$	$$250,000 \pm 75,000$
Operations and Maintenance per Year	\$41,080	\$130,700	78,904
Operations and Maintenance per 1,000 gallons treated	\$0.30 +/- 0.09	\$0.70 +/- 0.30	\$0.55 +/- 0.16

Acknowledgments

This document, the Arsenic Removal Demonstration Project Final Report, was prepared for the U.S. Environmental Protection Agency (EPA) National Risk Management Research Laboratory (NRMRL) in Cincinnati, Ohio, and the U.S. Department of Energy (DOE) Federal Energy Technology Center (FETC) in Pittsburgh, Pennsylvania, by MSE Technology Applications, Inc. (MSE) under contract DE-AC22-96EW96405. The Arsenic Removal Demonstration Project was conducted under the Mine Waste Technology Program (MWTP) funded by the EPA. The MWTP was jointly administered by EPA and DOE through an Interagency Agreement. MSE manages the MWTP and owns/operates the MSE Testing Facility in Butte, Montana, previously the DOE Western Environmental Technology Office.

Mr. Roger Wilmoth from NRMRL served as EPA's MWTP Program Manager, and Mr. Melvin Shupe from DOE served as DOE's Technical Program Officer. Mr. Creighton Barry served as MSE's Program Manager, Dr. Martin Foote served as MSE's MWTP Project Manager, and Mr. Jay McCloskey served as MSE's Technical Project Manager. Dr. Larry Twidwell from Montana Tech of the University of Montana (Montana Tech) and Mr. Glenn Vicevic from ZENON Environmental Inc. acted as technology providers and are recognized for their contributions. Dr. Twidwell was the developer of the Mineral-Like Precipitation Process and represented Montana Tech during the demonstration evaluation process. Mr. Vicevic represented ZENON, Inc. Both Dr. Twidwell and Mr. Vicevic provided engineering expertise before and during the demonstrations. In addition, both prepared demonstration reports for their respective technologies. The organization and execution of the MWTP Arsenic Removal Demonstration Project was a collaborative effort between the participants mentioned above.

In addition to the people listed above, the following agency and contractor personnel contributed their time and energy by participating in the Arsenic Removal Demonstration Project and preparing this document.

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Acronyms, Abbreviations, and Chemical Symbols

AA atomic absorption

AAM arsenic adsorption microfiltration

AHAP arsenatehydroxyapatite

ANSTO Australian Nuclear Science and Technology Organization

APHAP arsenatephosphate hydroxyapatite

As arsenic

As(III) arsenic(III), arsenite As(V) arsenic(V), arsenate

BDAT best demonstrated available technology

CCB continuing calibration blank CCV continuing calibration verification

cm centimeter COC chain-of-custody

CRC Cooperative Research Centre for Waste Management and Pollution Control Limited

DI deionized

DOE U.S. Department of Energy E_H oxidation-reduction potential

EPA U.S. Environmental Protection Agency

FCC factored capital cost

Fe iron

Fe(II) iron(II), ferrous Fe(III) iron(III), ferric

FETC Federal Energy Technology Center

FIT flow indicating transmitter

g grams

 $\begin{array}{lll} g/L & grams \ per \ liter \\ gpm & gallons \ per \ minute \\ H_2O_2 & hydrogen \ peroxide \\ HAP & hydroxyapatite \\ HCl & hydrochloric \ acid \\ HPDE & high-density \ polyethylene \end{array}$

IAG Interagency Agreement ICP inductively coupled plasma

ICP-AES inductively coupled plasma atomic emission spectroscopy

L liter

LCS laboratory control sample

M million

M&S Marshall and Swift
MDL method detection limit
mg/L milligrams per liter

mL milliliter

Montana Tech Montana Tech of the University of Montana

Acronyms, Abbreviations, and Chemical Symbols (cont.)

MSE Technology Applications, Inc.

MT/hr metric ton per hour

mV millivolts

MWTP Mine Waste Technology Program

NIST National Institute of Standards and Technology

NPV net present value

NRMRL National Risk Management Research Laboratory

ppb parts per billion ppm parts per million

psig pounds per square inch gauge

QA quality assurance

QA/QC quality assurance/quality control QAPP quality assurance project plan

QC quality control

RPD relative percent difference SOP Standard Operating Procedure

TCLP toxicity characteristic leaching procedure

TSS total suspended solids

XPS X-ray photoelectron spectrometry ZENON Environmental, Inc.

Fg/L micrograms/liter
Fm micrometer

1. Introduction

1.1 Project Management

This document is the Final Report for the U.S. Environmental Protection Agency's (EPA) Mine Waste Technology Program (MWTP) Activity III Project 9, Arsenic Removal Demonstration Project. The MWTP is a program developed through an Interagency Agreement (IAG) between EPA and the U.S. Department of Energy (DOE) (Ref. 1). MSE Technology Applications, Inc. (MSE) implements the MWTP and owns/operates the MSE Testing Facility in Butte, Montana. MSE proposed and was granted funding for the Arsenic Removal Demonstration Project during the December 1996 IAG Management Committee Meeting.

1.2 Project Purpose

The purpose of the Arsenic Removal
Demonstration Project was to demonstrate
alternative water treatment technologies
capable of effectively removing arsenic.
Several technologies with potential application
to treat water with arsenic problems were
presented in the MWTP Activity I, Volume 5,
Issues Identification and Technology
Prioritization Report—Arsenic (Ref. 2). Each
technology was screened and prioritized on the
basis of its potential to reduce arsenic levels
within arsenic containing waste streams.

1.3 Quality Assurance

The analytical methods and pilot-scale treatment testing conducted for this study were consistent with EPA's requirements outlined in the project-specific MWTP Activity III, Project 9 Quality Assurance Project Plan (QAPP) for the Arsenic Removal Project Demonstration (Ref. 3). The QAPP followed the EPA Category II procedures.

This final report describes the research that was conducted and summarizes the technical results that were obtained by evaluating the arsenic removal treatment technologies from mineral industry arsenic-bearing waters.

1.4 Technical Developers

1.4.1 Mineral-Like Precipitation

Mineral–like precipitation has been extensively investigated at the bench–scale by Dr. Larry Twidwell, a Montana Tech of the University of Montana (Montana Tech) professor. The research has been performed over a 10-year period on several different arsenic-bearing waters.

1.4.2 Alumina Adsorption with Microfiltration

Alumina adsorption is a widely recognized technology for the removal of arsenic from water. An innovative approach using alumina adsorption combined with microfiltration has been developed by ZENON Environmental, Inc. (ZENON) of Canada.

1.5 Scope of the Problem

Acidic, metal-bearing water draining from remote, abandoned mines has been identified by the EPA as a significant environmental/health hazard in the Western United States (Ref. 4). Many of these waters contain dissolved arsenic in the trivalent and pentavalent state.

Arsenic compounds and solutions are frequently an unwanted byproduct of the mining and metallurgical extraction of metals such as copper, gold, lead, and nickel. Arsenic waste problems will continue to grow as high-grade ores with low-arsenic content are depleted, and the processing of sulfide ores

with high arsenic content becomes increasingly common. An example of arsenic-bearing solid wastes from the processing of gold and base metal ores is the flue dust produced from roasting and smelting unit operations. The flue dust is often concentrated in arsenic; the arsenic is usually present as arsenic trioxide. Large quantities of flue dust from past and current mineral-processing operations are being kept in temporary storage pending the development of safe disposal methods.

The U.S. National Drinking Water Standard for arsenic is 50 parts per billion (ppb). Due to concerns for cancer risk associated with arsenic, the World Health Organization recently revised the guideline for arsenic in drinking water from 50 to 10 ppb in 1993 (Ref. 5).

Arsenic is a naturally occurring element commonly found in the mining industry. Dissolved arsenic has two common valence states (III and V). Generally, arsenic in the arsenite state (III) is more soluble than arsenic in the arsenate state (V). Due to this chemical trait, arsenic is generally removed more effectively from solutions in the oxidized or arsenate state (Ref. 6).

1.6 Statement of Project Objectives The primary objective of the field demonstration project was to assess the

effectiveness of the chosen processes for removal of arsenic from solution. Another objective of the project was to evaluate the products formed from each process to determine if they are environmentally stable. More specifically, the project objectives were:

- Reduction of the concentration of dissolved arsenic in the effluent waters to a level less than the National Primary Drinking Water Regulation Limit for arsenic established by the EPA of 50 ppb, or reduce the concentration of dissolved arsenic by 50% if the influent concentration was less than 50 ppb.
- C Production of the concentrated arsenic-bearing solids from the processes that are environmentally stable by demonstrating that arsenic results using TCLP will be below the maximum concentration for toxicity of 5.0 mg/L.

1.7 Demonstration Site Locations

A number of sites that have arsenic present in process or effluent streams were identified. Two sites selected for the demonstration were ASARCO Lead Smelter East Helena, Montana and TVX Mineral Hill Mine 1,300' Portal located in Jardine, Montana. Each site along with MSE in Butte, Montana is identified in Figure 1-1.

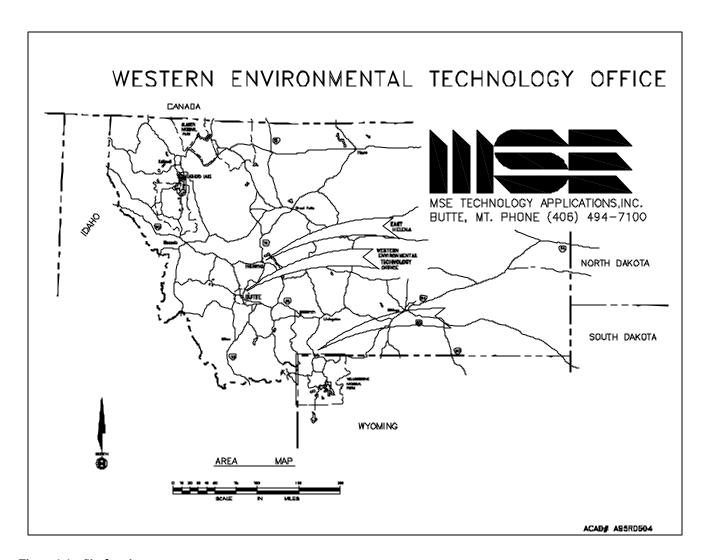


Figure 1-1. Site location map.

2. Project Organization and Responsibilities

2.1 Project Officers

Specific EPA, DOE, and MSE project officers and their respective responsibilities for Activity III, Project 9 are listed below.

EPA Project Officer—Roger Wilmoth: Responsible for EPA project management for MWTP and reviewing and approving the final project report.

DOE Project Officer—Mel Shupe: Responsible for DOE participation in the MWTP and reviewing and approving the final project report.

National Risk Management Research Laboratory (NRMRL)

Quality Assurance Associate—Kim McClellan: Responsible for reviewing and approving the QAPP.

MSE Program Manager—Creighton Barry: Responsible for senior review of all project plans and deliverables and for ensuring that the project objectives are achieved within schedule and budget constraints.

MSE MWTP Projects Manager—Martin Foote: Responsible for ensuring that the project is conducted according to the appropriate plans and that all project activities are documented in a project file. Also informs the Program Manager of the project status and of any technical/administrative/contractual/ financial issues and proposed resolutions.

MSE Arsenic Removal Technical Project Manager—Jay

McCloskey: Responsible for the execution of elements assigned

by the

Project Manager. Directly responsible for the

execution of field experiments and sampling schedule.

MSE Project Test Engineer—Dick Harned: Responsible for developing the test plan for the project.

Montana Tech Representative—Dr. Larry Twidwell:

Responsible for bench-scale testing of mineral-like precipitation process and scale up for the demonstration skid.

ZENON Representative—Glenn Vicevic: Responsible for construction and delivery of alumina adsorption skid to demonstration location, and input on experimental design related to ZENON skid.

MSE Technology Testing and Operations Manager—Vince

Tonc: Responsible for all aspects of testing and operations including safety and health and QA/QC.

MSE Project QA Officer—Helen Joyce: Responsible for developing the project QAPP, auditing test personnel and equipment and for submitting audit findings to the Technology Testing and Operations Manager, and independent data validation.

MSE-HKM Laboratory Manager—Kevin Kissell: Responsible

for ensuring that all analytical data meets quality objectives and for review of all laboratory reports. project and submitting findings to the QA Manager.

MSE-HKM Laboratory QA Officer—Jackie Timmer:

Responsible for reviewing all analytical data associated with the

3. Technology Descriptions

The three arsenic removal technologies demonstrated during the MWTP Activity III Project 9, Arsenic Removal Demonstration Project were 1) mineral–like precipitation; 2) alumina adsorption with microfiltration; and 3) ferrihydrite adsorption process.

At the request of the technology providers, minimal process information is provided in the following sections. Only sufficient information is provided to gain a basic understanding of each process.

3.1 Mineral-Like Precipitation of Arsenic

Mineral–like precipitation of arsenic from aqueous solution was investigated on a bench- scale level at Montana Tech by Dr. Larry Twidwell as part of MWTP Activity IV, Project 5—Removal of Arsenic from Waste Solutions as Storable Stable Precipitates (Ref. 7). The objective of this project was to strip arsenic from solutions in such a way so as to produce mineral-like precipitated products that are stable for long-term storage in outdoor pond environments. The approach investigated was the substitution of arsenate ions for phosphate ions in known phosphate minerals, such as hydroxy apatite [HAP, $Ca_{10}(PO_4)_6(OH)_2$] (Ref. 7). The mineral-like precipitation process is presented in Figure 3-1.

The MWTP Activity IV, Project 5 study resulted the following

positive results.

Arsenic can be effectively stripped to the parts per billion range from aqueous solutions by controlling the pH and P/As molar ratio in the initial solution.

A series of arsenatephosphate hydroxyapatite (APHAP)-bearing solid solutions can be formed by controlling the P/As molar ratio in the initial solution prior to precipitation. Solid solutions containing arsenic concentrations from approximately 3% to 30% have been formed. These are new compounds that have never been reported previously. The new compounds stoichiometry have been identified by chemical digestion and their structures by x-ray diffraction and x-ray photoelectron spectrometry (XPS).

The standard free-energy of formation of the APHAP compounds have been determined at 25E C. This information can be used to model the stability of the compounds under various solution conditions. One of the important considerations is whether the compounds will be stable for long-term storage in tailings pond environments (i.e., exposure to air). Previously, Dr. R.G. Robins had demonstrated that calcium arsenate compounds are unstable in air because the carbon dioxide in the air reacts with the calcium arsenate to form calcium carbonate and releases the arsenic back

to the solution phase (Ref. 8).

Modeling of tailings pond conditions shows that compound stability is a function of P/As mole ratio. Compounds with a P/As mole ratio greater than five should be stable to air exposure in tailings pond–type storage conditions.

Currently, compound stability is being tested by sparging air into

stability tests were performed on sludge products from this demonstration.

If the long-term stability of the solids formed using the mineral–like precipitation process is demonstrated, this process shows great promise for industrial applications. The mineral-like precipitation process is illustrated in Figure 3–1. Sampling locations for the mineral-like precipitation process are also shown in Figure 3–1 [101, 102, 104, 105, 106 Filter cake #1, pH, and flow indicating transmitter (FIT)]. The same skid was used for both the ASARCO and Mineral Hill Mine demonstrations. The skid was thoroughly decontaminated following the ASARCO demonstration before transportation to the MSE Testing Facility to treat the Mineral Hill Mine water. The pH elements indicate the location of pH probes to monitor the pH in tank number 101, tank number 102, and tank number 103. The FIT element in Figure 3-1 denotes the location of a flow totalizer.

3.2 Alumina Adsorption of Arsenic with Microfiltration

Alumina adsorption is a widely recognized technology for the removal of arsenic from water. An innovative approach of using alumina adsorption with microfiltration has been developed by ZENON. The arsenic adsorption microfiltration (AAM) process

aqueous/compound slurries. The pH, oxidation–reduction potential $(E_{\rm H})$, arsenic, phosphorus, and calcium concentrations are being monitored as a function of aging time.

After 6 months of aging, all dissolved arsenic concentrations remain below 50 Fg/L. Similar

is presented in Figure 3-2.

Arsenic–contaminated wastewater was pumped to the ZeeWeed Microfiltration process tank. The ZeeWeed Microfilter had a pore size of < 0.2 micrometers (μ m) and was installed directly in the process tank. Air was added to the module to continually move the fibers, thereby keeping them clean. The permeate was withdrawn from the process tank by applying a vacuum of 5 to 10 pounds per square inch gauge (psig) on the ZeeWeed membrane.

A suspension of finely divided activated alumina particles was charged to the ZeeWeed Microfiltration tank. The particles have an approximate size of 1.5 μm , and therefore, do not settle readily. It is the small size of the activated alumina that allows the AAM process to be effective because the surface area per particle accessible to the arsenic adsorption is extremely high as opposed to conventional alumina adsorbents that are used in columns. Therefore, the kinetics of adsorption are extremely favorable and rapid.

The first step of arsenic removal involves mixing the arsenic contaminated water with finely divided activated alumina in slurry form in an adsorption reactor. The wastewater was continually

pumped to the well-mixed ZeeWeed process tank and the arsenic was adsorbed onto the activated alumina. Bench-scale testing has demonstrated that the arsenic adsorption kinetics are favorable at a pH of 3 to 4. The ZeeWeed process tank was held at this pH using hydrochloric acid. The ZeeWeed membrane rejects the activated alumina particles, and the permeate (with a very low concentration of arsenic) was discharged. As more and more wastewater was processed, the activated alumina adsorption sites became occupied and the adsorbent was saturated. Regeneration of the absorbent was accomplished by the addition of sodium hydroxide to the process tank until the pH of the activated alumina was approximately 12. The arsenic was then desorbed from the alumina. Formation of a concentrated sodium arsenate brine was generated and recovered during the regeneration cycle. The brine was processed further to convert the arsenic to a physical and chemical form that was most suitable for offsite recycle, reuse, or disposal. Fresh wastewater or process effluent was fed to the process tank at the same rate as the permeate was withdrawn.

Once the alumina regeneration cycle was completed, the flow of the feed to the adsorption/regeneration tank was resumed for another treatment cycle. The concentrated sodium arsenate brine that was recovered during the alumina diafiltration was processed further to convert the arsenic to the physical and chemical form that was most suitable for offsite recycle, reuse, or disposal. The diafiltered solution was then directed to a conventional iron chloride coprecipitation process to recover the arsenic as a sludge. Sampling ports for the alumina adsorption with microfiltration skid are designated (301, 302, 304, 305, 306, FIT, and Filter cake #3) in Figure 3-2. The FIT element in Figure 3-2 denotes the locations of flow indicators. The skid was mobile to facilitate the setup at the different demonstration sites. The alumina adsorption with

microfiltration technology was demonstrated treating the Mineral Hill Mine 1,300 Portal water at the MSE Testing Facility in Butte, Montana, and the thickener overflow water at the ASARCO East Helena Smelter.

3.3 Ferrihydrite Adsorption of Arsenic

The ferrihydrite process is a commonly used industrial arsenic removal technique. This technology was used as the baseline technology for comparative purposes with the innovative technologies of alumina adsorption with microfilitration and mineral-like precipitation. The ferrihydrite process is illustrated in Figure 3-3.

In order for ferrihydrite adsorption to occur, the ferric ion (Fe⁺³) must be present in the water. Arsenic is most effectively removed from the water when oxidized to the arsenate (As⁺⁵) state and the Fe/As mole ratio is greater than 4 at a pH of 4 to 5. Dissolved arsenic is removed from the oxidated water by a lime neutralization process, in the presence of the Fe⁺³ which results in the formation of arsenic-bearing hydrous ferric oxide (ferrihydrite). The ferric ion is not stable in an aqueous environment above pH 7 and will precipitate out as ferric hydroxide (Ref. 9). The chemical reactions for these processes are listed below:

Formation of ferric hydroxide:

$$Fe^{+3} + 3H_2O -----> Fe(OH)_{3(s)} + 3H^+$$

Adsorption and coprecipitation of arsenic(V) with $Fe(OH)_{3(s)}$:

$$AsO_4^{!3} + Fe(OH)_{3(s)} ----->$$

$$Fe(OH)_{3(s)} + AsO_4^{-3}_{(ad)}$$

Acid neutralization with lime (CaO):

$$Ca(OH)_2 + 2H^+ ----> Ca^{+2} + 2H_2O$$

The ferrihydrite is separated from the treated water before the process of arsenic removal is complete. The solid-liquid separation is accomplished by a process involving conventional settling/flocculation with pressure filtration.

The ASARCO East Helena Lead Smelter had an existing ferrihydrite system that was used to compare removal efficiencies and process economics for the treatment of arsenic in water to the mineral–like precipitation process and alumina adsorption with microfiltration.

A pilot-scale ferrihydrite system was constructed to treat the Mineral Hill Mine water. Sampling ports are shown in Figure 3-3 for the Mineral Hill Mine demonstration (201, 202, 204, 205, 206, Filtercake #2, pH and FIT). The pH elements indicate the locations of pH probes in tank 201, tank 202, and tank 203. At the ASARCO East Helena Lead Smelter, the sampling ports were the influent, effluent, and sludge sampling locations currently used for the existing ferrihydrite system.

The arsenic-iron sludge from the Mineral Hill Mine Site Demonstration processes was expected to pass toxicity characteristic leaching procedure (TCLP). After this information was verified through testing, the sludge was disposed of at the Butte-Silver Bow Sanitary Landfill. The ASARCO sludge was recycled to its smelter operation.

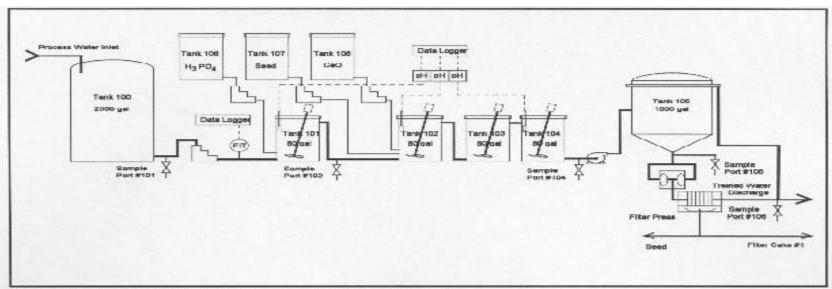


Figure 3-1. Mineral-like precipitation process flow diagram.

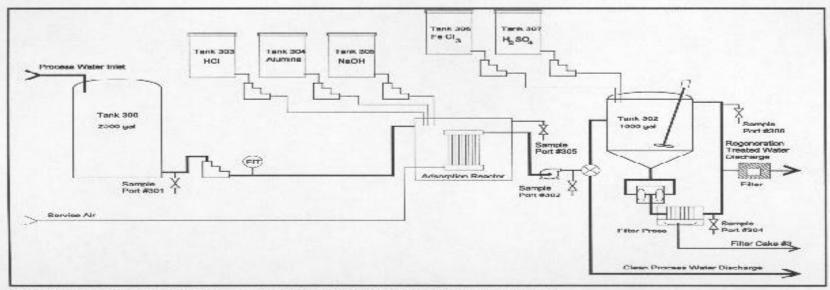


Figure 3-2. Alumina adsorption with microfilitration process flow diagram.

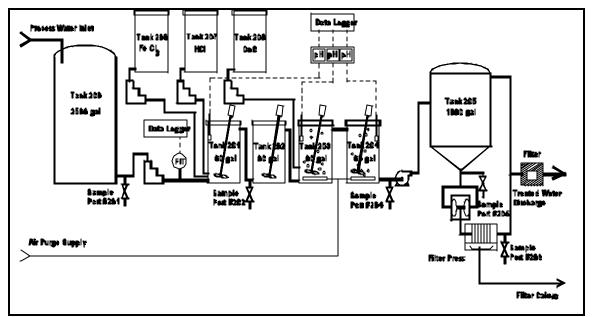


Figure 3-3. Ferrihydrite adsorption process flow diagram.

4. Site Descriptions

The pilot-scale demonstrations were performed at two sites: 1) ASARCO East Helena Lead Smelter in East Helena, Montana, and 2) MSE Testing Facility in Butte, Montana.

All field testing of these processes were conducted by MSE, Montana Tech, and ZENON personnel. Monitoring of pH, $E_{\rm H}$, and flow rates were performed at both sites.

All additional inorganic chemical analyses for samples collected at both sites are being conducted at the MSE-HKM Analytical Laboratory, which is located in Butte, Montana. Long-term stability tests are being conducted on the solids produced from the ferrihydrite process and the mineral-like precipitation process at Montana Tech.

4.1 ASARCO East Helena Lead Smelter

The ASARCO East Helena Lead Smelter has been in operation since the late 1800s. The East Helena Smelter is a custom, primary lead smelter that produces lead bullion from a variety of both foreign and domestic concentrates, ores, fluxes, and other nonferrous metal–bearing materials. The East Helena Smelter also produces byproducts such as silver, gold, copper, and sulfuric acid.

The East Helena Smelter is located within the City of East Helena, 3 miles east of Helena, Montana. The demonstration was housed

in the proximity of the Strike Clarified Acid Water Storage Tank Building and the technologies were tested using scrubber blowdown water and/or thickener overflow water from the gas cleaning system prior to entering the acid plant.

Analytical constituents for both the scrubber blowdown and thickener overflow water are presented in Tables 4-1 and 4-2, respectively.

4.2 Mineral Hill Mine

The demonstration was to be held at the Mineral Hill Mine. The Mineral Hill Mine is an underground gold mine owned by TVX Mineral Hill Mining, Inc., and had been in operation intermittently since World War II. Recently, mining operations at the Mineral Hill Mine ceased. The mine is located in Park County, at Jardine, Montana, 5 miles from the community of Gardiner, Montana.

For the duration of the demonstration, Mineral Hill Mine supplied the needed volume of water from the 1,300-foot level portal. Due to some logistical problems, it was decided to haul the Mineral Hill Mine 1,300' Portal water and perform the demonstration at the MSE Testing Facility in Butte, Montana. The untreated water was gravity fed to a tank truck and hauled to MSE. The treated water from the skid was analyzed and discharged to the Butte-Silver Bow sewer system. Analytical results for Mineral Hill Mine water can be seen in Table 4-3.

Table 4-1. Constituents of ASARCO Scrubber Blowdown Water

Constituent	Dissolved Species Concentration, µg/L
As(III)	3,913,000
As(V)	702,000
Cd	412,600
Ca	51,500
Cd	412,600
Cl	2,200,000
Cu	5,000
Fe	29,100
Pb	9,800
Mn	14,300
P	9,000
S	1,117,000
Se	10,400
Si	32,900
Zn	141,600

¹Dissolved concentrations (i.e., sample filtered through a 0.2 micron HDPE filter disk).

Table 4-2. Constituents of ASARCO Thickener Overflow Water

Constituent	Dissolved Species Concentration, µg/L
As(III)	4,060
As(Total)	5,810
Ca	732,400
Cd	20
Cu	10
Fe	30

Table 4-2. Constituents of ASARCO Thickener Overflow Water

Constituent	Dissolved Species Concentration, $\mu g/L$
Pb	< 20
Mn	20
P	24,600
S	812,000
Se	1,410
Zn	< 9

¹Dissolved concentrations (i.e., sample filtered through a 0.2 micron HPDE filter disk except for arsenic). The arsenic speciation was determined on an unfiltered sample.

Table 4-3. Constituents of Mineral Hill Mine 1,300' Portal Water

Constituent	Dissolved Species Concentration, $\mu\text{g}/L$
As(V)	366
As(Total)	362 (range was 366-670)
Ca	124,600
Cd	10
Cu	10
Fe	< 24
Pb	< 20
P	< 30
SO_4^{-2}	181,000
Zn	< 10

¹Dissolved concentration (i.e., sample filtered through a 0.2 micron HPDE filter disk).

5. Quality Assurance

MSE developed a QAPP (Ref. 3) to control the experimental test work design for the Arsenic Removal Project. The Test Plan (Ref. 9) was developed to implement the QAPP and the associated scope of work for the demonstrations and management of each project. The QAPP was written to followed EPA Category II procedures. Sampling

Procedures, and Analytical Procedures and Calibration along with sample port/location descriptions, sample matrix, noncritical and critical measurements and frequency for each process demonstration are attached in Appendix A. For further information on the experimental design, refer to the QAPP and Work Plan references (Refs. 3 and 9).

6. Field and Laboratory Data Validation Report

In August of 1997, sampling officially began for the MWTP Activity III, Project 9—Arsenic Removal Demonstration. Sampling, analyzing, and calibration procedures are presented in Appendix A.

6.1 Project Audits

An audit of a specific portion of each technology demonstration was performed throughout the project:

- C Field Systems Audit at ASARCO for the Alumina Adsorption Process;
- C Metals Analysis Review at Montana Tech for the Mineral–Like Precipitation Process; and
- C Sample Collection/Decontamination Procedures Review at the MSE Testing Facility for the Ferrihydrite Adsorption Process.

6.1.1 Field Systems Audit at ASARCO

A field systems audit was performed on August 14, 1997, at the Alumina Adsorption process demonstration at the ASARCO Lead Smelting Plant. The Alumina Adsorption process was demonstrated by ZENON Environmental, Inc. The system audit included a review of the following items:

- personnel, facilities, and equipment;
- documentation [chain-of-custody (COC), logbooks];
- calibration of equipment; and
- sampling procedures.

No concerns were identified during the audit.

6.1.1.1 Personnel, Facilities, and Equipment

Personnel present during the audit included: Jay McCloskey, Technical Project Manager; Dave Reisenauer, Operator; Glenn Vicevic, ZENON, Project Manager; and Greg McGinn, ZENON, Operator.

The demonstration was held at the ASARCO lead smelting plant, located in East Helena, Montana. Equipment for the demonstration was housed in the acid plant at the smelter. Analysis and preparation of the samples (filtering, preserving) was performed in the plant on a table specifically designed for that purpose. Project personnel were knowledgeable about the demonstration and their duties and responsibilities at the demonstration site.

All ZENON equipment was checked prior to shipment with National Institute of Standards and Technology (NIST) secondary standards on a scheduled basis. All calibration information was available on the equipment checkout sheet provided by ZENON. ZENON is ISO and its personnel were familiar with quality assurance (QA) procedures in general.

6.1.1.2 Documentation

Chain-of-custody procedures were reviewed at the demonstration site and all COC procedures were being followed. The project logbooks were also reviewed. The sampling logbook was very thorough and included spaces where specific information was required. Sampling personnel were familiar with the logbook format and COC procedures.

6.1.1.3 Calibration of Equipment

Field equipment was used to collect pH and flow rate. This information was recorded from digital readouts on the demonstration skid. The pH was also verified by collecting a

sample and measuring the pH with a pH meter. Standard operating procedures (SOP) were available at the demonstration site to calibrate/operate the pH meter and the ZENON alumina adsorption skid. Sampling personnel were familiar with the SOPs and requirements for routine calibration of the pH meter.

6.1.1.4 Sampling Procedures

A review of sampling activities was also performed during the systems audit. Operations personnel from MSE and ZENON were trained by MSE-HKM Laboratory personnel in proper sampling procedures. ZENON personnel were familiar with sampling procedures because similar procedures were used during the ZENON demonstration for the Resource Recovery Project at the MSE Testing Facility. All sample collection and equipment decontamination procedures were followed by sampling personnel.

Samples were stored in a refrigerator at 4 °C at the demonstration site prior to shipment to the laboratory. The most critical holding times were for arsenic and iron speciation. Samples were shipped via ground transportation in sealed coolers filled with ice by project personnel. The drive from the ASARCO lead smelting plant to the laboratory took approximately 60 minutes. For the mineral-like precipitation process demonstration, an audit of arsenic analysis was performed at Montana Tech.

6.1.2 Metals Analysis Review at Montana Tech

In addition to the systems audit in the field, an audit of Inductively Coupled Plasma (ICP) with Hydride Generation for arsenic analysis in the Metallurgy Department at Montana Tech was performed on the evening of August 14, 1997. The analyst was Michelle Gale, a graduate student from Montana Tech assisting Dr. Larry Twidwell on the mineral-like precipitation demonstration. For the entire duration of the project, ICP analysis had been performed by Montana Tech to assess how the process is operating and allow for process changes if warranted. The

purpose of the metals analysis at Montana Tech was to determine the arsenic concentration throughout the process with shorter turnaround times than the MSE-HKM Laboratory. Arsenic was the only analysis performed.

A Varian Liberty 110 ICP was used for the analysis. The ICP was calibrated with three standards and a blank. Quality control (QC) checks included continuing calibration verification (CCV), continuing calibration blank (CCB), preparation blank, laboratory control sample (LCS), and duplicates. During the first run, the ICP analysis was out of control limits for the CCV. A sampler tube was changed, the ICP was recalibrated, and the analysis proceeded until all of the samples had been analyzed. The analysis generated some interesting results that identified a possible problem with decontamination of tanks and hoses following the ASARCO demonstration of the mineral-like precipitation process. The influent to the system has a concentration of approximately 450 ppb, while the initial tanks in the system had concentrations of approximately 600 ppb. These results indicated that arsenic was being added to the system rather than being removed. When the effluent sample was analyzed, the result was only 8 ppb, which easily met the objective. To determine the source of the arsenic contamination, the decontamination procedures following the ASARCO demonstration were most likely not rigorous enough. While the tanks were acid washed, it was discovered that the hoses may not have been flushed thoroughly, which could account for the elevation of arsenic in the early stages of the process. In any case, the mineral-like precipitation process was able to remove the additional arsenic. In the future, the wastewater with the lowest concentration (Mineral Hill Mine Water) should be demonstrated first and then the wastewater with the higher concentration (ASARCO) could be demonstrated.

6.1.3 Sample Collection/ Decontamination Procedures Review at the MSE Testing Facility

For the ferrihydrite adsorption of arsenic portion of the demonstration, sample collection procedures were witnessed on 09/05/97, near the end of the demonstration. The purpose of the audit was to oversee sample collection and equipment decontamination procedures performed by project personnel at the MSE Testing Facility. Rich Henningsen, a process engineer, performed the sampling. First, a sample for iron and arsenic was collected from the ferrihydrite adsorption process, then filtered using pressure filtration, preserved with nitric acid, capped, labeled, and recorded in the project logbook. The filtering apparatus was then decontaminated with a 1:1 nitric acid and thoroughly rinsed with deionized (DI) water. To determine whether decontamination procedures were effective, a field external decontamination blank was collected. The field external decontamination blank results give an indication of contamination introduced through sampling procedures, field equipment (filter and filtering apparatus), preservation, and carryover after decontamination, as well as contamination introduced in the laboratory. The MSE-HKM Laboratory reported the results of this sample, and the results showed no contamination at < 40 ppb by inductively coupled plasma atomic emission spectroscopy (ICP) and < 1 ppb by atomic absorption (AA) analysis. The results of this blank indicate that the decontamination procedures for sampling equipment used during the project were rigorous enough. Results of other field QC samples are discussed in Section 6.2.4 of this report.

6.2 Data Evaluation

In addition to the systems audits performed during the project, all field and laboratory data has been evaluated to determine the usability of the data. The final project samples were collected on September 8, 1997.

To determine the effectiveness of the arsenic removal processes being demonstrated, several sampling points were designated for each process and a variety of analyses were assigned to each point. The analyses to be performed were specified in the project-specific QAPP (Ref. 3), and each analysis was classified as critical or noncritical. A critical analysis is one that must be performed in order to achieve project objectives. A noncritical analysis is one that is performed to provide additional information about the process being tested.

Critical analyses for this project are summarized below.

- Dissolved arsenic: and
- TCLP for arsenic.

Noncritical analyses for this project are listed below:

- pH;
- E_{H} ;
- total flow;
- temperature;
- flow rate:
- arsenic speciation;
- iron speciation;
- dissolved metals (Al, As, Cd, Cu, Fe, Pb, P, Zn, Ca);
- total recoverable metals (Al, As, Cd, Cu, Fe, Pb, P, Zn, Ca);
- total metals (As, Ba, Cr, Cd, Cu, Fe, Pb, P, Ag, Zn, Ca);
- percent solids; and
- TCLP (Ba, Cd, Cr, Pb, Hg, Se, Ag).

The QC objectives for each critical analysis were outlined in the QAPP and were compatible with project objectives and the methods of determination being used. Additional information on critical and noncritical analysis is available in Appendix A. The QC objectives are method detection limits (MDLs), accuracy, precision, and completeness. Control limits for each of these

objectives were established for each critical analysis. For noncritical analyses, QC objectives were determined using standard guidelines that exist or applying reasonable control limits in order to determine the usability of the data.

6.3 Validation Procedures

Data that was generated for all critical and noncritical analyses was validated. The purpose of data validation is to determine the usability of all data that was generated during the project. Data validation consists of two separate evaluations: 1) an analytical evaluation, and 2) a program evaluation.

6.3.1 Analytical Evaluation

An analytical evaluation is performed to determine the following:

- C All analyses were performed within specified holding times;
- C Calibration procedures were correctly followed by field and laboratory personnel;
- C Laboratory analytical blanks contain no significant contamination:
- C All necessary independent check standards were prepared and analyzed at the proper frequency and all remained within control limits:
- C Duplicate sample analysis was performed at the proper frequency, and all Relative Percent Differences (RPDs) were within specified control limits;
- Matrix spike sample analysis was performed at the proper frequency and all spike recoveries (%R) were within specified control limits; and

C Data in the report submitted by the laboratory to project personnel can be verified from the raw data generated by the laboratory.

Measurements that fall outside of the control limits specified in the QAPP, or for other reasons are judged to be outlier, were flagged appropriately to indicate that the data is judged to be estimated or unusable. All QC outliers for all sampling events are summarized in Table 6-2. In addition to the analytical evaluation, a program evaluation was performed.

6.3.2 Program Evaluation

Program evaluations include an examination of data generated during the project to determine the following:

- C All information contained in COCs is consistent with the sample information in field logs, laboratory raw data, and laboratory reports;
- C All samples, including field QC samples, were collected, sent to the appropriate laboratory for analysis, and analyzed and reported by the laboratory for the appropriate analyses;
- C All field blanks contain no significant contamination; and
- C All field duplicate samples demonstrate precision of field as well as laboratory procedures by remaining within control limits established for RPD.

Program data that was inconsistent or incomplete and did not meet the QC objectives outlined in the QAPP were viewed as program outliers and were flagged appropriately to indicate the usability of the data. Both the analytical and program evaluations consisted of evaluating the data generated in the field as well as in the laboratory.

6.4 Analytical Evaluation

The analytical evaluation of field and laboratory data was completed in November of 1997.

6.4.1 Field Logbook Evaluation

Field data validation began with an examination of the field log books that were created for this project. Sampling logbooks were created for each process test. General site logbooks were also created for the demonstration by MSE personnel and Australian Nuclear Science and Technology Organization (ANSTO) personnel. The field logbook typically contains all of the information that is available regarding the following:

- information about fieldwork performed; and
- sample collection activities,

6.4.1.1 Information About Fieldwork Performed

The general logbooks contained daily logs of fieldwork performed and process measurements taken. Feed and tank changes were noted in the general logbook.

6.4.1.2 Sample Collection Activities

Sampling logbooks contained all of the appropriate information for sample collection and field measurements that were taken. Sampling conditions and information such as weather conditions, date of sampling, time of sampling, and details of fieldwork performed should be specified in the field logbook for each sampling event. Sampling information was complete and accurate for all sampling events. While a specific space was not provided for additional comments or information, sampling personnel made notes in the margins when necessary. The sampling logbook format facilitated review by specifying a space for each

measurement to be recorded in; therefore, missing information was easy to locate. All of the preservatives required for each analysis were clearly listed in the sampling logbooks. The logbooks for the alumina adsorption process contained the wrong sample preservatives for sulfate and total suspended solids (TSS) analysis; however, this problem was discovered and corrected in the field although it could have been a serious problem that affected the entire test series for the alumina adsorption process. Table 17 in the QAPP summarized the analyses and the proper preservatives for all of the analyses and similar tables in future QAPPs should be consulted when creating logbooks. In addition, any new personnel used for sampling activities should be made aware of the QAPP and the procedures outlined in the QAPP to avoid this mistake in the future.

6.4.2 Field Data Validation

Field data validation was performed to determine the usability of the data that was generated during field activities. The usability was determined by verifying that correct calibration procedures of field instruments were followed. Standard operating procedures for calibration of field instruments were available at the demonstration site. All of the field measurements were classified as noncritical. The following measurements were performed in the field:

- E_{H} ;
- pH;
- temperature;
- flow rate; and
- total flow.

Table 6-1 summarizes the measurements that were not recorded in the logbook for the various tests. There was no justification provided in the logbook as to why the data was not collected. The reason(s) for not recording measurements should be provided in the logbook.

$6.4.2.2 E_{H}$

An Orion E_H meter with a silver/silver chloride reference electrode was used to determine the E_H of samples at the demonstration site. The electrode was calibrated using Zoebell's solution of known E_H . All E_H data are considered usable. The calibration was not documented in the project logbooks; however, during the demonstration audit, project personnel indicated the E_H meter was calibrated each day measurements were taken. All equipment calibrations should be documented in the project logbooks. Refer to Table 6-1 for the dates and times E_H was not recorded in the logbook for each test.

6.4.2.3 pH

The pH meter was calibrated using two known buffer solutions that would bracket the measured pH. Calibration of the pH meter was performed each day pH measurements were taken. The pH data were also recorded from pH meters installed in the process skids, if available. All pH data are considered usable. Refer to Table 6-1 for the dates and times pH was not recorded in the logbook for each test.

6.4.2.4 Temperature

Temperatures of the process inlets and outlets were measured using the thermistor in the pH meter or mercury thermometers. The thermistor and the mercury thermometers were calibrated by the Instrumentation and Control Department on a regular basis according to the manufacturer's instructions. All temperatures were recorded in the project logbooks when measurements were taken, and all temperature data is considered usable. Refer to Table 6-1 for the dates and times temperature measurements were not recorded in the logbook for each test.

6.4.2.5 Flow Rate and Total Flow

Flow rates and total flow were determined using flow meters or flow totalizers installed within each process skid. Project personnel recorded the flow rate or total flow from a digital readout. Refer to Table 6-1 for the dates and times that flow measurements were not recorded in the project logbooks.

6.4.3 Laboratory Data Validation

Laboratory data validation was performed to determine the usability of the data that was generated by the laboratory for the project. The following analyses were performed in the MSE-HKM Laboratory:

- arsenic speciation (noncritical);
- iron speciation (noncritical);
- dissolved metals (Al, As, Cd, Cu, Fe, Pb, P, Zn, Ca) (critical and noncritical);
- TCLP (critical);
- total recoverable metals (Al, As, Cd, Cu, Fe, Pb, P, Zn, Ca, Na) (noncritical);
- total metals (As, Ba, Cd, Cr, Cu, Fe, Pb, P, Zn, Ca, Na) (noncritical); and
- percent solids (noncritical).

Laboratory data validation was performed using *USEPA Contract Laboratory Program National Functional Guidelines for Inorganics Data Review* (Ref. 10) as a guide (where applicable) to each individual analysis. For critical analyses, the QC criteria outlined in the QAPP were also used to identify outlier data and determine the usability of the data for each analysis. When data validation was initiated, the MSE-HKM Laboratory was not sending sufficient information to perform a complete and thorough data validation. Due to the large volume of data generated for the project, the data validation was performed at the laboratory or by

electronic copy rather than requiring the laboratory to submit copies of all data generated for the project.

6.4.3.1 Arsenic Speciation

Arsenic speciation analysis was performed at the MSE-HKM Laboratory. The concentration of As⁺³ and As⁺⁵ in project samples was determined using furnace AA, following the speciation of the arsenic using the Ion Exchange Ficklin Method. The procedure involves passing 5 milliliters (mL) of the filtered, acidified sample through an ion exchange column packed with Donwex 1 x 8 anion-exchange resin in 100-200 mesh size. The As⁺⁵ adheres to the acetate form of the ion exchange resin while the As⁺³ passes through the column. To ensure the recovery of all of the As⁺³, the column is eluted with three separate 5-mL portion of DI water. The original sample and each elution are collected in separate vials numbered 1 through 4. These vials contain the As⁺³ from the original sample. The column is then eluted with three separate 5 mL portions of 0.12 million (M) hydrochloric acid (HCl). The pH change and the subsequent ion exchange causes the As⁺⁵ to pass through the column and the three vials containing the last three elutions contain the As⁺⁵. All of the speciation vials as well as an unspeciated total dissolved arsenic sample will be analyzed by furnace AA to determine the concentrations of As⁺³, As⁺⁵, and total dissolved As. Samples requiring qualification for arsenic speciation analysis are summarized in Table 6-2.

6.4.3.2 Iron Speciation

Iron speciation was performed at the MSE-HKM Laboratory. The concentration of ferrous iron will be determined using a modified colorimetric Standard Method 3500-Fe D from *Standard Methods for the Examination of Water and Wastewater*, which uses phenanthroline as the color developer. Total iron is measured similarly after reducing the iron in the sample to the ferrous state by boiling using acid and hydroxylamine and treating with 1,10-phenanthroline at a pH of 3.2 to 3.3. The concentration of the

ferric iron was then calculated by subtracting the concentration of ferrous iron from the concentration of total iron. The spectrophotometer was calibrated with a blank and at least three standards. All iron speciation data is considered usable and required no qualification.

6.4.3.3 Aqueous Metals Analysis by ICP

Dissolved and total recoverable metals concentrations and concentrations in TCLP extracts were determined using SW-846 Method 6010A on a Varian Liberty 110 ICP. The samples were prepared according to SW-846 Method 3005A. The ICP was calibrated according to procedures outlined in SW-846 Method 6010A and the equipment manufacturer's instructions. Calibration consisted of the following procedures and items:

- mixed calibration standards:
- calibration blanks and reagent blanks;
- independent check standards;
- interference check solutions; and
- QC samples.

Refer to Table 6-2 for samples requiring qualification for ICP analysis.

6.4.3.4 Aqueous Metals Analysis by Atomic Adsorption

Because the ICP was not sensitive enough to detect arsenic concentrations below 40 ppb, all samples with concentrations below 100 ppb by ICP were reanalyzed by AA to more accurately determine the concentration of arsenic at lower levels. One batch of AA analyses required qualification due to an analytical spike that was out of control. Because the spike recovery was low (76.5%), there may be a slight negative bias in the arsenic concentrations for these samples. Refer to Table 6-2 for the data requiring qualification for AA analysis.

6.4.3.5 Toxicity Characteristic Leaching Procedure

Solid materials were subjected to the TCLP procedure outlined in SW-846 Method 1311 at the MSE-HKM Laboratory. The resulting extraction fluids from the TCLP were digested according to procedures outlined in SW-846 Method 3005A for total recoverable metals. In addition to the reagents listed in the method, 20 mL of 30% hydrogen peroxide (H_2O_2) was added to the samples prior to digestion to help degrade the acetic acid. Digested samples were analyzed by ICP with the exception of mercury. Mercury was analyzed by cold vapor AA according to SW-846 Method 7470. All TCLP data is considered usable, and none of the TCLP data was qualified.

6.4.3.6 Solid Metals Analysis/Percent Solids

Solid samples were characterized for total metals by ICP according to SW-846 Method 6010A at the MSE-HKM Laboratory. Samples were digested according to SW-846 Method 3050A. The percent solids of each sample was also determined using the method outlined in Exhibit D, Part F of the *Contract Laboratory Program Statement of Work*, Document Number IL03.0. The method involves weighing a wet portion of the solid material, drying the sample in an oven to constant weight, then reweighing the sample to determine the moisture lost during drying. The percent solids data was used to report the total metals on a dry weight basis. All percent solids data is considered usable and required no qualification.

6.5 Program Evaluation

The program evaluation focused on the following areas:

- COC procedures;
- sampling and data completeness;
- field blanks; and
- field duplicates.

6.5.1 Chain-of-Custody Procedures

All information provided in the COC forms for this project was complete and accurate; however, on several occasions changes were made on the COC forms and were not initialed and dated. All changes made to COC or project logbooks should be made by striking out the mistake with a single line and initialing and dating the change.

6.5.2 Sampling and Data Completeness

All samples that were supposed to be collected were collected. During several tests extra samples were also collected and fully documented in the project logbooks. All collected samples were analyzed for the requested analyses on the COC forms.

6.5.3 Field QC Samples

All field QC samples were collected at the proper frequency for tests specified in the QAPP. All samples requiring qualification due to field QC samples are summarized in Table 6-2.

6.5.3.1 Field Blanks

None of the field blanks collected for the project showed significant contamination.

6.5.3.2 Field Duplicates

Field duplicates showed very good agreement with the original samples, with the following exception:

Alumina Adsorption Test at ASARCO

Dissolved arsenic duplicate sampled on 08/14/97 was out of control for arsenic analysis by AA.

See Table 6-2 for a summary of qualified data due to this out of control field QC sample.

6.6 Summary

While the majority of the findings of the analytical and program evaluations are minor and can be easily addressed or have already been addressed, several lessons can be learned so that mistakes will not be repeated during future projects. The following recommendations are suggested in order to improve future project and program QA/QC.

6.6.1 Laboratory QA/QC

 $\mathrm{QA/QC}$ summaries and raw data were available for review at the MSE-HKM Laboratory upon

request; however, prior to future projects, project personnel should inform any laboratory performing analyses about QA/QC reporting needs (QA/QC summaries and raw data should be attached to the report).

6.6.2 Field QA/QC

Field sample identification numbers included the sample port number so that influent and effluent samples could be distinguished from one another on the laboratory report. When questionable data was reported, the reviewer had to have the field log data sheets to determine if the sample was an influent or effluent. More descriptive sample identification numbers would make the data review process much easier.

There was a great volume of data generated during this project, and while some of the data is considered estimated for various reasons, the fact that all of the data is usable underlines the fact that quality data was generated for MWTP Activity III, Project 9.

Table 6-1. Summary of Field Measurements Not Recorded in Logbook

Time	Measurements not recorded
at ASARCO	
21:35, 00:30, 1:29	Flow rate
Mineral Hill Mine Water	
22:00 (time was recorded as 10:00; however, military time should be used so each sample time is unique to avoid confusion)	Flow rate
10:00	рН
16:00	Total Flow
Mineral Hill Water	
9:20, 13:20	Temperature
07:00	Flow rate, Temperature, pH, E_{H}
Similar data should be collected on extra samples for comparability	pH
	at ASARCO 21:35, 00:30, 1:29 Mineral Hill Mine Water 22:00 (time was recorded as 10:00; however, military time should be used so each sample time is unique to avoid confusion) 10:00 16:00 Mineral Hill Water 9:20, 13:20 07:00 Similar data should be collected on extra samples

Table 6-2. Summary of Qualified Data for MWTP Activity III, Project 9

Date ¹	Sample ID	Analysis	QC Criteria	Control Limit	Result	Flag ²	Comment
	ASARCOLL#	As Speciation	Total Recovery	80%-120% Recovery of Arsenic Species	125.9% Recovery	J	The recovery of arsenic species was outside specified control limits.
	ASARCOLL# 2	As Speciation	Total Recovery	80%-120% Recovery of Arsenic Species	123.2% Recovery	J	The recovery of arsenic species was outside specified control limits
	Feed 1A	Total Recoverable Cu Fe Pb Al Mg P	Duplicate	#20% RPD	23% RPD 22.6% RPD 48.4% RPD 43% RPD 21.8 % RPD 47.1% RPD	J J J J	Duplicate results differed significantly enough to flag associated samples "J", as estimated.
08/14/97	PLA2-105 PLA2-106 PLA2-108 PLA2-110 PLA2-111 PLA2-113 PLA2-115 PLA2-118 PLA2-126	Diss As by AA	Field Duplicate	Reviewer Discretion (Laboratory Duplicate Control Limit is #20% RPD)	63.8% RPD	J	Field duplicate results differed significantly enough to flag associated samples "J", as estimated.
08/18/97	ALA-162	Diss As Al	Duplicate	#20% RPD	136.2% RPD 56.6% RPD	J J	Duplicate results differed significantly enough to flag associated samples "J", as estimated.
08/17/97 08/17/97	MLM-344 MLM-346	Diss P	CCV	89-111% Recovery	112.3% Recovery	J	Samples over the IDL are qualified "J" as estimated due to out of control CCV.
08/17/97 08/17/97 08/17/97	ALA-132A ALA-133A ALA-132B	Diss Al	CCV	89-111% Recovery	112.3% Recovery	J	Samples over the IDL are qualified "J" as estimated due to out of control CCV.

Table 6-2. Summary of Qualified Data for MWTP Activity III, Project 9 (cont.)

Date ¹	Sample ID	Analysis	QC Criteria	Control Limit	Result	Flag ²	Comment
08/04/97 08/04/97 08/04/97 08/04/97 08/03/97 08/03/97 08/04/97 07/24/97 07/24/97 07/25/97	MLA-214 MLA-216 MLA-216A MLA-216B MLA-202 MLA-204 MLA-108 MLA-110A MHA-102 MHA-104A MHA-119A	Diss As by AA	Analytical Spike	85%-115% Recovery	76.5%Recovery	J	Analytical spike results were out of control and the sample concentration was less than 50% of the spike concentration.

MHA-120

MHA-108 MHA-108A

MHA-124 MHA-125

MHA-126

07/25/97

07/24/97

07/24/97 07/25/97

07/25/97 07/25/97

¹ Date that the samples were collected.

² Data Qualifier Definitions:

U—The material was analyzed for, but was not detected above the level of the associated value (quantitation or detection limit).

J—The sample results are estimated.

R—The sample results are unusable.

UJ—The material was analyzed for, but was not detected, and the associated value is estimated.

¹ Date that the samples were collected.

² Data Qualifier Definitions:

U-The material was analyzed for, but was not detected above the level of the associated value (quantitation or detection limit).

J-The sample results are estimated.

R-The sample results are unusable.

UJ-The material was analyzed for, but was not detected, and the associated value is estimated.

7. Demonstration Results

7.1 Mineral-Like Precipitation Results

7.1.1 ASARCO Scrubber Blowdown Water Analytical Results

The removal of arsenic from ASARCO scrubber blowdown water by the mineral–like precipitation technology using a phosphorous to arsenic mole ratio of 7 was very effective. The effluent water from the settler (after 24-hours continuous operation) contained < 10 ppb arsenic (the goal was to remove the arsenic to below 50 ppb). The experimental results are summarized in Table 7-1, while the complete experimental results are presented in Appendix B. The influent composition and the final effluent water from the treatment system are presented in Table 7-2. The solution pH, $E_{\rm H}$, and temperature data are summarized in Table 7-3.

The removal of arsenic from scrubber blowdown water is dependent on the addition of the proper amount of lime (see Figure 7-1). The solubility of arsenic as APHAP is depicted in this figure. The data used to generate the figure is based on the standard free energy of formation of APHAP as determined by Twidwell, et al (Ref. 7). Note that approximately 40 g/L lime should be required for effective removal of arsenic for an influent arsenic concentration of 3.0 g/L. The demonstration test was conducted using 56 g/L lime. Figure 7-2 shows that approximately 50 g/L lime was required to raise the pH of the scrubber blowdown water to 11 and above.

7.1.2 Solid Characterization

Solids were collected from the settler at the end of each test run. The percent solids in the settler bottoms were 21.9% (for the 24-hour test) and 20.2% (for the 3-hour test). Aliquot samples were split from the slurries and were

used for settling rate studies. Each settler bottom slurry was filtered. The filtrates were saved and used in the long-term stability tests (see Section 7.1.4) and will be used to replace the solution that evaporates with time from the long-term aging samples. The solids were saved for characterization studies that included elemental characterization, x-ray diffraction, scanning electron microscopy (and energy dispersive spectroscopy), and long-term stability during storage.

7.1.3 Toxicity Characteristic Leaching Procedure

The TCLP was performed on the composite solids produced at the end of each treatment series. Total metals concentration and TCLP results are presented in Table 7-4 for the 24-hour, P/As mole ratio = ~ 5.5 test and Table 7–5 for the 3-hour, phosphorus (P)/As mole ratio = ~ 11.9 test. Note that product solids from the 24-hour test passed the TCLP test but the solids from the short (higher phosphorus) time (3-hour) test did not. Therefore, the first test solids are considered to be nonhazardous with respect to handling and land disposal; however, the second test solids are considered hazardous. The reason that the second test solids did not pass the TCLP is presently not known, although the very short duration of the test (i.e., only 3 hours) may be the reason. Even though the second test solids did not pass the TCLP test, the solids are being subjected to long-term leach testing.

7.1.4 Long-Term Stability

The need for long-term stability testing was previously presented in Section 3.1 [i.e., the solids that are produced by other technologies (lime precipitation and ferrihydrite adsorption) may not be stable for long-term outdoor storage]. The mineral-like precipitation

technology solves the storage problem because the product is thermodynamically stable against conversion to calcium carbonate by carbon dioxide in atmospheric air.

To validate that the mineral-like product was indeed stable, long-term stability tests were initiated and will be continued for 2 years. Briefly, the aging test procedure consists of the following steps.

- C One-hundred grams of filter press solids (percent moisture determined) were placed in 1-liter of effluent solution in high-density polyethylene (HDPE) bottles. Triplicate test slurries were prepared.
- C The slurries were shaken for 24 hours, pH and $E_{\rm H}$ were determined, and solution samples were extracted, preserved, and submitted to MSE-HKM for analytical characterization. These samples are considered time zero for the aging demonstration.
- C Each sample bottle was then set so that air could be sparged into the slurry at 10 mL/min. Presently, solution pH, $E_{\rm H}$, and temperature are being monitored monthly.
- C The solubility of the solids will be determined after 1 and 2 years of exposure.

The time zero analytical results for the ASARCO scrubber blowdown water are presented in Table 7-6.

7.1.5 X-ray Diffraction

The solids were subjected to x-ray diffraction. The x-ray diffraction patterns are presented in Figure 7-3. The patterns for the solids produced from both the 20-hour and 3-hour tests appear to be very similar. The pattern for

the 20-hour test is presented in Figure 7-3. Note that a semicrystalline product is represented and that there is a cluster of peaks in the 2-theta range 30-36E. This pattern is similar (but the crystallinity is not yet well developed) to the apatite and apatite-like minerals. The APHAP compounds have the same crystal structure as hydroxyapatite (HAP) and arsenatehydroxyapatite (AHAP) (i.e., Johnbaumite). This is illustrated in Figure 7-4. Note that the solid solution APHAP compound major peaks lie between the HAP (no As present) and AHAP (no P present) major peaks. Note also that the pattern for the ASARCO solid (which contains approximately 2.1%-2.7% arsenic) as seen in Figure 7-3, when superimposed on the pattern for APHAP (which contains 2.9% arsenic) shows excellent similarity (see Figure 7-5).

7.2 ASARCO Thickener Overflow Analytical Results

The removal of arsenic from ASARCO thickener overflow water by the mineral-like precipitation technology was very effective [e.g., the effluent water from the settler (after 8-hours continuous operation) was less than 15 ppb arsenic (the goal was to remove the arsenic to below 50 ppb)]. The experimental results are summarized in Table 7-7. The input water composition and the final effluent water composition from the treatment system is presented in Table 7-8. The solution pH, E_H, and temperature data are summarized in Table 7-9. The arsenic removal was enhanced by increasing the P/As mole ratio. The removal of arsenic was very rapid at the higher P/As ratio (i.e., the arsenic content was less than 10 ppb in less than 15 minutes). This effect is demonstrated by the data presented in Table 7-9.

The data for two separate tests are summarized in Table 7-7 (i.e., the first test was conducted for 16 hours using a nominal

P/As mole ratio of 10; the second test was conducted for 20 hours using a nominal P/As mole ratio of 100). The reason for increasing the P/As mole ratio to 100 is described below.

Samples were taken from Tank 101 (the phosphate addition tank) early in the test period. These samples were taken to Montana Tech for quick analyses. The results showed there was essentially no phosphorus available in the solution phase in Tank 101 (i.e., samples filtered through 0.2 µm filter disks showed only a few parts per billion phosphorus present but samples not filtered showed the proper phosphorus content). The phosphorus was being adsorbed onto an organic phase (probably from the filteraid used in the thickener). Therefore, the first test was terminated at 16 hours. However, as noted in Table 7-7, the loss of phosphorus from the aqueous solution turned out to be a nonissue [i.e., excellent arsenic removal was achieved (after approximately 8 hours of operation) at the lower P/As ratio].

The removal of arsenic from ASARCO water treatment thickener overflow water requires only minor lime addition. It would appear that the thickener water (already at a pH of ~ 11 and a calcium content of ~ 730 mg/L) should precipitate the APHAP compound without addition of more lime. However, as the data from sampling port 102, (the water exiting the inlet reactor plus phosphate addition tank) shows, arsenic was not precipitated (see Figure 3-1). The reason for this result is because much of the phosphate was adsorbed onto the entrained flocculant. However, by adding more lime to the system, arsenic was stripped from the solution and flocculant. The demonstration test was conducted by adding sufficient lime to bring the lime content in the water to 1 g/L. This addition, illustrated in Figure 7-6, raised the solution pH to approximately 12. It is likely that effective

arsenic removal could have been achieved with a much smaller lime addition (i.e., note that in Figure 7-6, the solution pH would still be above 10, even at a lime addition rate of 0.25 g/L). The unknown at this point is whether minor lime addition rates would strip the phosphate from the flocculate.

7.2.1 Toxicity Characteristic Leaching Procedure

The TCLP was performed on the composite solids produced at the end of each treatment process. Total metals concentration and TCLP results are presented in Table 7-10 for the 20-hour test, P/As mole ratio= 100. Note that product solids from the treatment sequence passed the TCLP test. Therefore, these solids are considered to be nonhazardous with respect to handling and land disposal.

7.2.2 Long-Term Stability

The need for long-term stability testing was presented previously in Sections 3.1 and 7.1.4, [i.e., the solids produced by other technologies (lime precipitation and ferrihydrite adsorption) may not be stable for long-term outdoor storage]. The mineral-like precipitation technology solves the storage problem because the product is thermodynamically stable against conversion to calcium carbonate by carbon dioxide in atmospheric air.

To validate that the mineral–like product is indeed stable, long–term stability testing was initiated and will be continued for 2 years. The experimental test procedure was presented in Appendix A. Briefly, the aging test procedure consists of the following steps:

C One hundred grams of filter press solids (percent moisture determined) were placed in 1 L of effluent solution (in HDPE bottles). Triplicate test slurries were prepared.

- C The slurries were shaken for 24-hours, the pH and $E_{\rm H}$ were determined, and solution samples were extracted, preserved, and submitted to MSE-HKM for analytical characterization. These samples are considered time zero for the aging demonstration.
- C Each sample bottle was placed so that air could be sparged into the slurry at 10 mL/min. Solution pH, E_H, and temperature are presently being monitored monthly.
- C The solubility of the solids will be determined after 1 and 2 years of exposure.

The time zero analytical results are presented in Table 7-11.

7.2.3 X-ray Diffraction

The solids from the thickener overflow water were subjected to x-ray diffraction analysis. The results showed that solids formed were similar to the scrubber blowdown water solids. The x-ray diffraction patterns for the thickener overflow water solids and the scrubber blowdown water are superimposed in Figure 7-7.

7.3 Mineral Hill Mine 1,300' Portal Results

The removal of arsenic from Mineral Hill Mine 1,300' Portal groundwater by the mineral-like precipitation technology was very effective [e.g., the effluent water from the settler (after only 1 hour of continuous operation) was < 10 ppb (one sample was 25 μ g/L, the goal was to remove the arsenic to below 50 ppb)]. The experimental results are summarized in Table 7-12. The input water composition and the final effluent water from the treatment system is presented in Table 7-

13. The solution pH, E_H , and temperature data are summarized in Table 7-14.

The removal of arsenic from Mineral Hill Mine groundwater is dependent on the addition of the proper amount of lime, see Figure 7-8. The solubility of arsenic as APHAP is depicted in this figure. The data used to generate the figure are based on the standard free energy of formation of APHAP, as determined by Twidwell, et al (Ref. 7). Note that less than 0.10 g/L lime should be required for effective removal of arsenic.

The demonstration test was conducted using three different lime addition rates shown on the diagram in Figure 7-9 [i.e., the treatment started at 1 g/L lime (for 32 hours), was subsequently decreased to 0.5 g/L (after 32 hours), then was decreased to 0.25 g/L for the reminder of the demonstration]. Also note in Figure 7-9 that the solution pH was still above 10 even at a lime addition rate of 0.25 g/L (solids must be formed at a pH of 10 or greater to ensure that the product is stable for long–term storage).

7.3.1 Batch Tests

Residence time, effect of P/As mole ratio, and effect of hydrated lime content were determined in a series of large–scale batch tests. The procedure and experimental results are presented below.

7.3.1.1 Residence Time

The residence time was determined by flowing process solution into a single reactor and measuring the arsenic concentration as a function of fill time. Two tests were conducted; one at a flow rate of 1 gallon of groundwater/minute and the second at a flow rate of 2 gallons of groundwater/minute. The following parameters were held constant for both tests: P/As mole ratio was 20, and the hydrated lime concentration was 0.5 g/L. The

results are presented in Table 7-15. The arsenic content was lowered to below the project goal (< 50 ppb) in less than 15 minutes residence time.

7.3.1.2 Effect of P/As Mole Ratio

Reactor vessels 102, 103, 107 were used to conduct 75-gallon batch tests. Each vessel was filled with Mineral Hill Mine groundwater and phosphoric acid was added to give P/As mole ratios of 10, 20, and 200. Hydrated lime was then added to the three vessels (each agitated) at the same time, and samples were collected as a function of time. The experimental results are presented in Table 7-15. The experimental results show that the higher the P/As mole ratio in the starting water, the lower the achievable arsenic content in the treated water. However, all the mole ratios investigated showed arsenic removal from the solution to below 50 µg/L in less than 5 minutes of reaction time.

7.3.1.3 Effect of Hydrated Lime Content

Reactor vessels 102, 103, 107 were used to conduct 75-gallon batch tests. Each vessel was filled with Mineral Hill Mine groundwater and phosphoric acid was added to provide a P/As mole ratio of 10. Hydrated lime was then added to the three vessels (each agitated) at the same time and samples were collected as a function of time. The experimental results are presented in Table 7-16. The experimental results show that hydrated lime concentrations between 0.1 g/L and 0.5 g/L are required. The large-scale continuous test demonstrated effective arsenic removal at a lime content of 0.2 g/L.

7.3.1.4 Solid Characterization

Solids were collected from the settler at the end of the test run. The percent solids in the settler bottoms were $1.2\pm0.1\%$. One liter of the aliquot samples was split from the slurries and used for settling rate studies (see Section

7.3.3). Each settler bottom slurry was filtered. The filtrates were saved and used to set up the long-term stability tests (see Section 7.2.3) and replace the solution that evaporates with time from the long-term aging samples. The solids were saved for characterization studies, including elemental characterization, x-ray diffraction, scanning electron microscopy (and energy dispersive spectroscopy), and long-term stability during storage.

7.3.1.5 Toxicity Characteristic Leaching Procedure

The TCLP was performed on the composite solids produced at the end of the treatment series. Total metals concentration and TCLP results are presented in Table 7-17. Note, product solids from the treatment sequences passed the TCLP test; therefore, these solids are considered to be nonhazardous with respect to handling and land disposal.

7.3.1.6 Long-Term Stability

The need for long-term stability testing was presented previously in Sections 3.1 and 7.1.4, the solids that are produced by other technologies (lime precipitation and ferrihydrite adsorption) may not be stable for long-term outdoor storage. The mineral-like precipitation technology solves the storage problem because the product is thermodynamically stable against conversion to calcium carbonate by carbon dioxide in atmospheric air.

To validate that the mineral-like product is indeed stable, long-term stability testing was initiated and will be continued for 2 years. The experimental test procedure was presented in Appendix A. Briefly, the aging test procedure consists of the following steps:

One hundred grams of filter press solids (percent moisture determined) were placed in 1-liter of effluent solution (in HDPE bottles). Triplicate test slurries were prepared.

- C The slurries were shaken for 24-hours. The pH and $E_{\rm H}$ were determined and solution samples were extracted, preserved, and submitted to MSE-HKM for analytical characterization. These samples are considered time zero for the aging demonstration.
- C Each sample bottle was then set up so that air could be sparged into the slurry at 10 mL/min. Solution pH, E_H, and temperature are presently being monitored monthly.
- C The solubility of the solids will be determined after 1 and 2 years of exposure.

The time zero analytical results are presented in Table 7-18. For comparison the ferrihydrite technology results are also presented in Table 7-18. Note that the mineral-like precipitation is equal to, or more effective for removing all the quoted elements, especially arsenic.

7.3.2 Settling Rate

The envisioned utilization of the mineral-like precipitation technology is that the precipitated solids will be clarified in a thickener. The overflow water will be the discharge water. The underflow slurry from the thickener will be pumped to a storage pond. The separation of solids requires a properly sized thickener. Therefore, a preliminary evaluation of settling rate was conducted using the Kynch method. Refer to the MWTP Activity III, Project 9—Mineral-Like Precipitation Studies by Dr. Larry Twidwell for a description of the Kynch method and the results of the settling rate tests (Ref. 11).

The required thickener size is approximately 1,154*M (square meters of thickener surface area, where M is the solids flow rate in metric tons per hour). A sizing exercise is presented below for a contaminated Mineral Hill Mine water feed rate of 300 gallons per minute (gpm) containing 500 ppb arsenic. This sizing exercise is based on a settling data that did not utilize any flocculant. Flocculant addition would enhance the settling rate.

The process will produce 0.00078 metric ton per hour (MT/hr) of product solids at the assumed water flow rate [P/As mole ratio= 10, lime requirement 1.5 times the stoichiometric requirement for $Ca_{10}(As_{0.11}P_{0.89}O_4)_6(OH)_2$].

A thickener of 1154 m²/MT/hr*0.00078 MT/hr = 0.9m² would therefore be required.

The diameter of the thickener would be 1.1 m (3.5 ft).

The required thickener diameter would be 3.5 ft. The smallest industrially available thickener is 4 ft (diameter). Solid/liquid separation does not appear to be a problem.

7.3.3 X-ray Diffraction

The x-ray diffraction pattern for the product from the Mineral Hill Mine 1,300' Portal groundwater is presented in Figure 7-10. This pattern shows that the product is primarily calcium carbonate. The arsenic content is $\sim 0.02\%$, therefore, the APHAP in the demonstration product would not be expected to be seen by x-ray diffraction.

7.4 Alumina Adsorption Results

Four tests were conducted treating ASARCO thickener overflow water and one test treating Mineral Hill Mine 1,300' Portal water. Complete analytical results for the alumina adsorption tests are presented in Appendix B.

7.4.1 ASARCO Pilot Analytical Results

7.4.1.1 Feed Water Arsenic and pH

During the demonstration using alumina adsorption with microfiltration, it should be noted that dissolved arsenic concentration varied. Examples of this variation can be observed in Figures 7-11, 7-12, and 7-13. Consequently, none of the data in Appendix B for the ASARCO tests was qualified in MSE–HKM's data validation report.

There was an inverse correlation between pH and arsenic concentration. As pH decreased from thickener overflow water, arsenic concentration increased.

Dissolved arsenic in aqueous solution exists as the arsenite and arsenate salts, which are highly soluble over a wide pH range with the exception of the calcium salt. The decrease in arsenic solubility with increased pH was likely due to adsorption of the arsenite and arsenate anions onto suspended particles in the feed water, and/or coprecipitation of these anions with other species. The average pH of the ASARCO feed water samples was 7, while the average ZeeWeed process tank was 4. Since lower pH is related to a higher dissolved arsenic content. it is likely that there was further dissolution of the arsenic when the feed water entered the lower pH process tank. The activated alumina was likely exposed to a higher dissolved arsenic concentration than feed water analysis would suggest, and pilot data must be evaluated on the basis of final permeate quality, rather than percent removal basis.

7.4.1.2 ASARCO Feed Water Arsenic Speciation

Results from the two arsenic speciations performed on oxidized ASARCO water are

shown in the Appendix B. Analysis of the first sample taken shows no oxidation, while analysis of the second sample shows complete oxidation of arsenic from + 3 to the + 5 state.

7.4.1.3 Trial 1 Test 1

Dissolved arsenic and pH of the feed and permeate are plotted in Figure 7-11. There are two anomalous dissolved arsenic analyses for the permeate. The dissolved arsenic concentration for the permeate sample was measured at 4.3 ppm. It was hypothesized that the high arsenic level of this sample was due to poor adsorption onto activated alumina since the ZeeWeed was measured at 2 ppm for pH 4, the dissolved arsenic content of the permanent sample should have been near this level or lower because the process tank pH was 7.6. However, the dissolved arsenic content for the permeate sample was measured at 3.16 ppm at a pH of 3.9. Since the dissolved aluminum analysis for this permeate sample was high, there was possible contamination of the sampling equipment from the TSS sampling event.

The average dissolved arsenic content of the permeate was 1.42 ppm at an average pH of 4.1. The average dissolved arsenic measured in the feed was 1.46 at an average pH of 4.7.

For permeate sample number PLA1-148A, the dissolved arsenic analysis was 2.28 ppm, while the total arsenic was 2.38 ppm. All samples analyses for the alumina adsorption tests are provided in Appendix B. These two values are in close agreement (5% RPD), indicating that only dissolved arsenic passes through the ZeeWeed membrane.

7.4.1.4 Diafiltration 1a

Dissolved arsenic and pH of the feed and the permeate from the Diafiltration Trial 1 Test 1 can be observed in Figure 7-12. The dissolved arsenic analysis of 2.55 ppm at a pH of 8.9 for

sample number is high since the preceding sample from the same feed tank has a dissolved arsenic analysis of 0.4 ppm at a pH of 8.3.

The average dissolved arsenic concentration in the feed during the diafiltration was 0.09 ppm at pH 12. The average permeate dissolved arsenic concentration was 0.18 ppm at pH 12, higher than the dissolved arsenic in the feed, as expected during diafiltration. However, it should be noted that the permeate dissolved arsenic concentration was 0.219 ppm at the start of diafiltration, whereas it was 2.3 ppm at the end of the adsorption trial. The lower concentration of dissolved arsenic in the sample would seem to indicate that arsenic had desorbed from the activated alumina during recirculation at pH 12, and the dissolved arsenic in the process tank had coprecipitated with or adsorbed onto some other compound at the high pH of diafiltration and therefore, had not passed through the membrane during diafiltration. Since the total arsenic concentration in the ZeeWeed process tank during diafiltration is not known, the amount of arsenic desorbed at high pH cannot be calculated.

7.4.1.5 Trial 2 Test 1

Dissolved arsenic and pH of the feed and permeate from Trial 2 of Test 1 are plotted in Figure 7-13.

During this trial, the dissolved arsenic content of the permeate stream rose continuously, while the pH was relatively constant at 3.9. The amount of arsenic that exited in the permeate during this trial was calculated at 40,800 mg. This value was checked against the total amount of arsenic in the system. Calculated on the basis of dissolved arsenic in the feed, the amount of arsenic that had been introduced to the system from the beginning of the test to the end of Trial 2, Test 1 was only

27,400 mg, which is less than the amount of dissolved arsenic that exited the system. This further supports the hypothesis that some of the arsenic that was nonsoluble in the feed dissolved in the low pH environment of the ZeeWeed process tank.

Arsenic that was desorbed from the alumina during preceding diafiltration coprecipitated with, or was adsorbed onto another precipitate at the high pH of diafiltration, and therefore, could not exit in the permeate during diafiltration. When the pH was dropped for Trail 2 of Test 1, the coprecipitated/adsorbed arsenic gradually redissolved but did not adsorb onto the alumina. As the arsenic redissolved, it passed through the ZeeWeed membrane and exited in the permeate.

7.4.1.6 Diafiltration 1b

Dissolved arsenic concentration and pH of the feed and the permeate from the diafiltration can be observed in Figure 7-14. The average dissolved arsenic concentration in the feed was 0.088 ppm at pH 11.8, and the average dissolved arsenic concentration in the permeate was 0.318 ppm at pH 12. The dissolved arsenic was higher in the permeate than in the feed during diafiltration. The dissolved arsenic concentration in the permeate after recirculation of the process tank at pH 12 was lower than the dissolved arsenic concentration in the permeate at the end of the preceding adsorption step. The first permeate sample of the diafiltration had a concentration of 0.315 ppm dissolved arsenic, while the last permeate sample in the preceding adsorption step had a dissolved arsenic concentration of 7.5 ppm. This suggests that the high pH of the diafiltration step caused both the arsenic that had adsorbed from the alumina and the arsenic in the process tank volume to come out of solution.

7.4.1.7 Test 2

Dissolved arsenic concentration and pH of the feed and permeate from Test 2 is shown in Figure 7-15. The dissolved arsenic in the first Test 2 permeate sample was higher than the dissolved arsenic in the permeate at the end of the preceding diafiltration, and it increased in the subsequent sample. This suggests that the arsenic had precipitated out in the ZeeWeed process tank at pH 12 during diafiltration, redissolved, and passed through the membrane at the lower pH of Test 2. The alumina concentration was increased to 30 g/L. Dissolved arsenic in the permeate was approximately 270 ppb at an average pH of 3.9. A higher concentration of alumina was successful in reducing the dissolved arsenic concentration in the process tank from its previous level.

The average sulfate concentration was 2,160 ppm in the feed and 1,730 ppm in the permeate indicating that some of the sulfate adsorbed onto the alumina. Consequently, the capability of the alumina to adsorb arsenic was reduced.

7.4.1.8 Test 3

Dissolved arsenic and pH of the feed and permeate from Test 3 are plotted in Figure 7–16. Diafiltration was not performed between Tests 2 and 3, and the purpose of running Test 3 was to determine how the membrane would perform at a high solids level (60 g activated alumina per L). The membrane performance was reported above in the description of the data in Figure 7-16.

Throughout Test 3, the feed water dissolved arsenic analysis fluctuated depending on the pH of the feed water, increasing as the pH increased.

The average dissolved arsenic concentration of the permeate in Test 3 was 183 ppb at an average pH of 4.2. The average sulfate concentration was 2,330 ppm in the feed and 1,790 ppm in the permeate. The arsenic adsorption capacity of the alumina may have been reduced by adsorption of sulfate.

7.4.1.9 Test 4

Dissolved arsenic and pH of the feed and permeate can be observed in Figure 7-17. The feed water for Test 4 was not treated with KMnO₄ for oxidation of As(III) to As(V).

The average dissolved arsenic concentration in the feed was 963 ppb at a pH of 10.3. The average dissolved arsenic concentration in the permeate was 334 ppb at a pH of 4.1.

The average sulfate concentration was 2,500 ppm in the feed and 1,940 ppm in the permeate, and the capability of the alumina to adsorb arsenic may have been reduced due to the adsorption of sulfate.

7.4.2 Mineral Hill Mine Water

7.4.2.1 Pilot Analytical Results

Dissolved arsenic and pH of the feed and permeate from the test on Mineral Hill Mine Water can be seen in Figure 7-18. The average dissolved arsenic concentration in the feed was 446 ppb at a pH of 4.2. The arsenic in the feed was completely dissolved (total and dissolved analyses were within 5% RPD). The average dissolved arsenic concentration in the permeate was 21 ppb at a pH of 3.8.

The average sulfate concentration was 236 ppm in the feed and 162 ppm in the permeate. The sulfate concentration in the Mineral Hill Mine water was much lower than in the ASARCO water, and the amount of sulfate adsorbed per gram of alumina was lower in the Mineral Hill Mine test.

7.4.2.2 Diafiltration

Dissolved arsenic and pH of the feed and permeate is shown in Figure 7-19. The dissolved arsenic concentration in the permeate was higher during diafiltration than during the preceding adsorption phase, indicating there was some desorption of arsenic from the activated alumina. The total amount of arsenic introduced to the ZeeWeed process tank was approximately 9,900 mg, based on the dissolved arsenic analysis of the feed. The total amount of arsenic that exited in the permeate during diafiltration was approximately 430 mg.

The dissolved arsenic in the permeate rose at the beginning of the diafiltration and then declined steadily although the feed water had a higher arsenic concentration that the process tank. The reason for the low initial concentration is not known. One hypothesis for the steady decrease is that there was precipitation of other species at the high pH of the process tank (pH 11.7) and arsenic which had desorbed from the alumina subsequently adsorbed onto these precipitated species.

7.5 Ferrihydrite Adsorption

7.5.1 ASARCO Analytical Results

For removal of arsenic in ASARCO thickener overflow water, two separate parameters were used. An iron to arsenic mole ratio of both 8 and 10 was used for ferrihydrite adsorption. The removal of arsenic from thickener overflow water is dependent on the amount of iron inputted into the system. Ferrihydrite adsorption technology was performed at the East Helena site using their current existing facility.

7.5.1.1 Low Iron Demonstration

Using an iron to arsenic mole ratio of 8, arsenic concentrations were lowered from 6.3 ppm to 100 ppb at pH of 7. Ferrihydrite adsorption was effective for arsenic removal, however, the established drinking water standard of 50 ppb was never achieved at this iron to arsenic mole ratio.

The analytical results can be observed in Table 7-19.

7.5.1.2 High Iron Demonstration

Increasing the iron content was very effective for removal of arsenic from ASARCO thickener overflow water. Concentrations were lowered from 6.3 ppm to less than 20 ppb. The input water composition and the final effluent results are summarized in Table 7-20.

7.5.2 Mineral Hill Mine 1,300' Portal Water

7.5.2.1 Analytical Results

For Mineral Hill Mine water, a pilot-scale process was constructed for the ferrihydrite adsorption technology.

The removal of arsenic using Mineral Hill Mine water by ferrihydrite adsorption was very effective. Results indicate arsenic concentrations were lowered from 600 ppb to less than drinking water standards of 50 ppb. The arsenic to iron mole ratio used for this demonstration was 10, which proved to be sufficient. Results can be seen in Table 7-21. Complete analytical results for the ferrihydrite adsorption tests are provided in Appendix B.

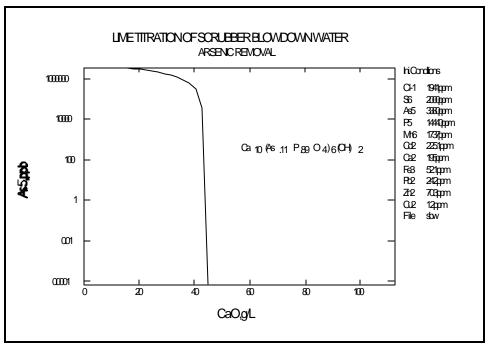


Figure 7-1. Lime titration of scrubber blowdown water: arsenic removal as a function of added lime.

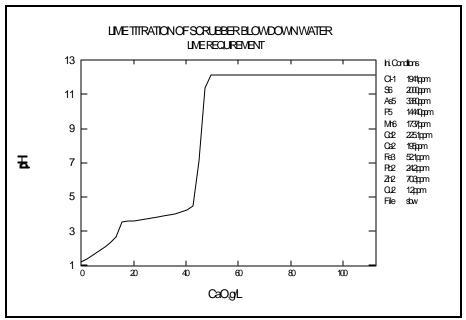


Figure 7-2. Lime titration of scrubber blowdown water: pH as a function of added lime.

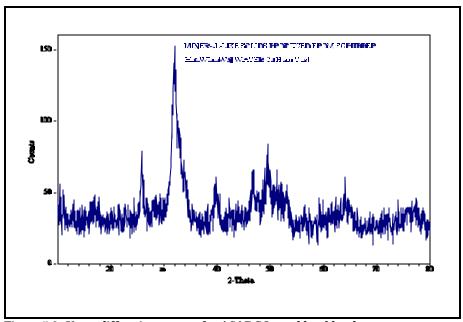


Figure 7-3. X-ray diffraction pattern for ASARCO scrubber blowdown water.

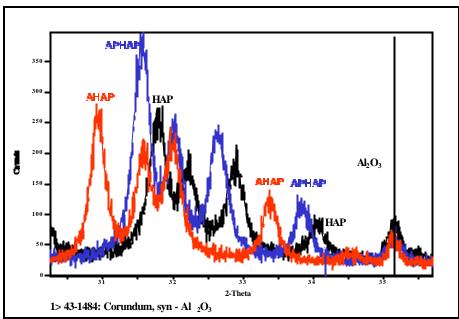


Figure 7-4. X-ray diffraction patterns for HAP, AHAP, and APHAP.

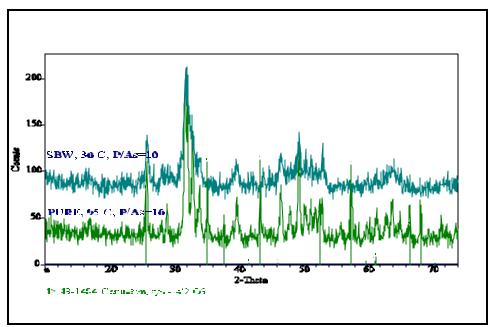


Figure 7-5. X-ray diffraction pattern for ASARCO scrubber blowdown water solid product superimposed on APHAP (both containing approximately 2-3% arsenic)

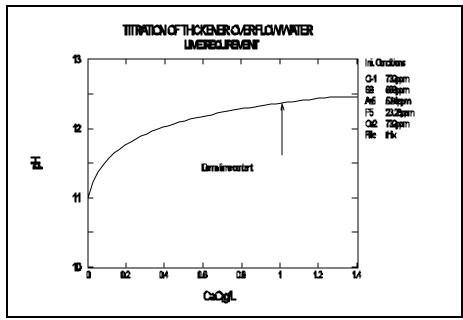


Figure 7-6. Hydrated lime titration of ASARCO water treatment thickener overflow water: pH as a function of added lime.

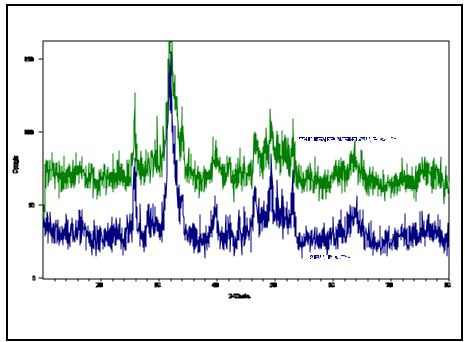


Figure 7-7. X-ray diffraction pattern for ASARCO thickener overflow water solid product superimposed on scrubber blowdown water solid product.

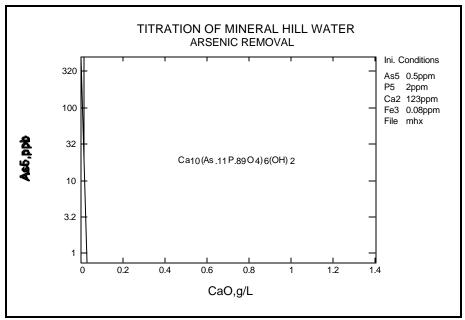


Figure 7-8. Hydrated lime titration of Mineral Hill Mine 1,300' Portal groundwater: arsenic removal as a function of added hydrated lime.

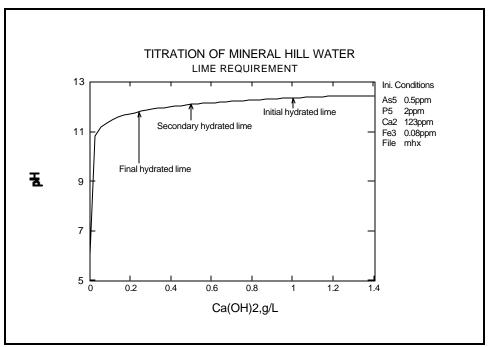


Figure 7-9. Hydrated lime titration of Mineral Hill Mine 1300' Portal groundwater: pH as a function of added hydrated lime.

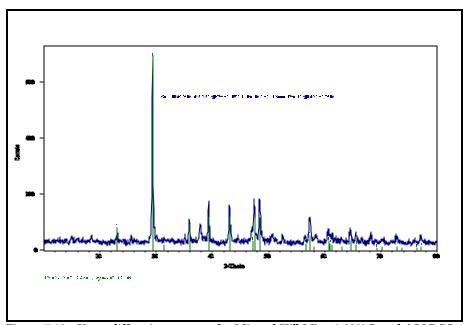


Figure 7-10. X-ray diffraction patterns for Mineral Hill Mine 1,300' Portal ASARCO thickener overflow water solids.

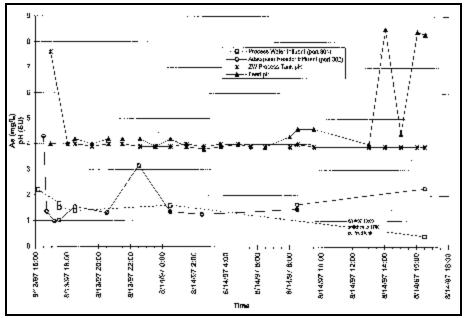


Figure 7-11. ASARCO-Trial 1 of Test 1 (activated alumina at 5 g/L). Dissolved arsenic and pH.

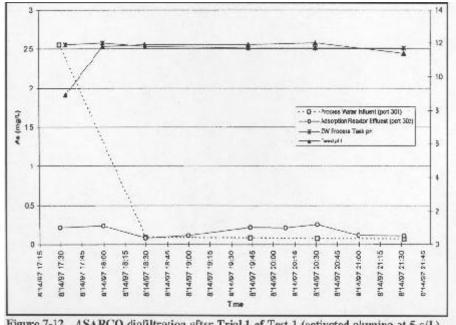


Figure 7-12. ASARCO dialiltration after Trial 1 of Test 1 (activated alumina at 5 g/L). Dissolved arsenic.

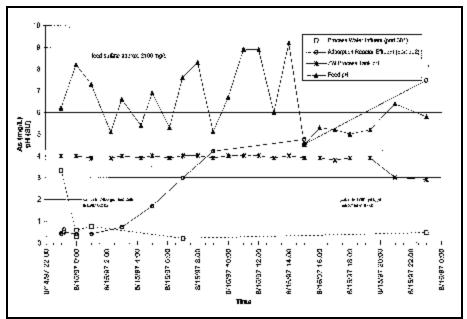


Figure 7-13. ASARCO—Trial 2 of Test 1 (activated alumina at 5 g/L). Dissolved arsenic and pH.

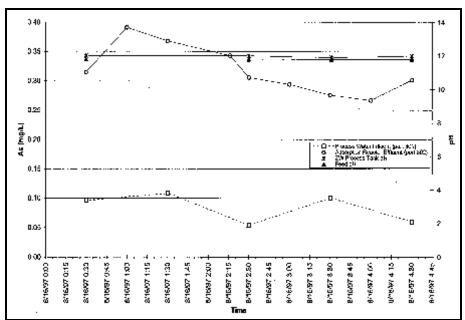


Figure 7-14. ASARCO diafiltration after Trial 2 of Test 1 (activated alumina at 5 g/L). Dissolved arsenic and pH.

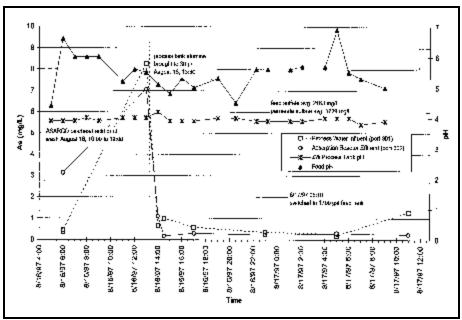


Figure 7-15. ASARCO Test 2 (activated alumina at 30 g/L). Dissolved arsenic and pH.

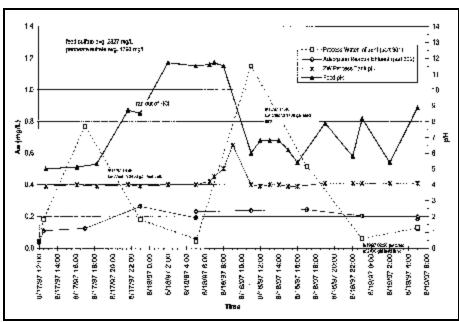


Figure 7-16. ASARCO Test 3 (activated alumina at 60 g/L). Dissolved arsenic and pH.

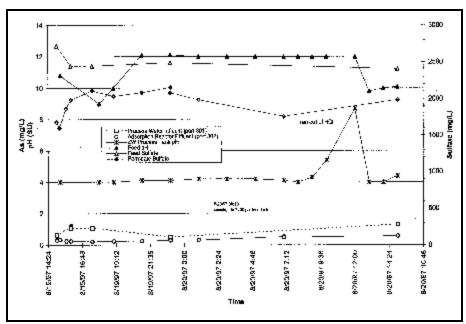


Figure 7-17. ASARCO Test 4 (activated alumina at 20 g/L noKMnO $_4$). Dissolved arsenic and pH.

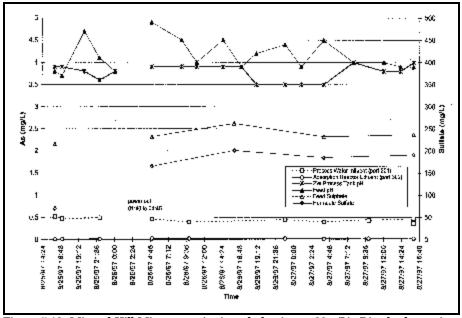


Figure 7-18. Mineral Hill Mine water (activated alumina at 20 g/L). Dissolved arsenic and pH.

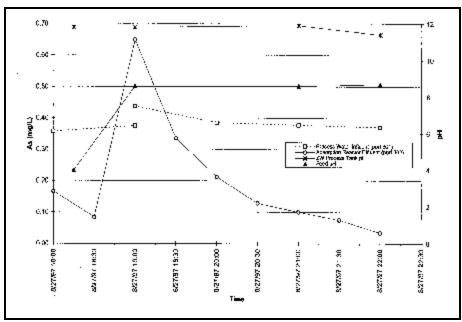


Figure 7-19. Diafiltration after Mineral Hill Mine water (activated alumina at 20 g/L). Dissolved arsenic and pH.

Table 7-1. Summary of Laboratory Test Results (P/As Mole Ratio=7)

Conditions [As], μg/L Average V1 V2 V3 Test1 Residence Solution pH General Initial Final Composite Time/Vessel Hours [As] Hours [As] [As] Hours min. 1 1.7 1 2.0 5.1 (14 hrs) Seed 3.4% into vessel-1. recovery= 4 8.2 4 1.7 Min. Hill Slurry feed rate 21 cc/min. 12.4 38 454 98.9% Solution feed rate 5 cc/min. (Test BS 3) 5.2 8 3.2 P/As mole ratio 7000². 14 2.2 1 41.8 1 29.311.7 (16 hours) Seed 1.6% into vessel-1. recovery= 4 26.2 4 12.9 Min. Hill Slurry feed rate 20.0 cc/min. 39 12.4 468 97.4% (Test BS 5) Solution feed rate 10 cc/min. 8 21.2 8 4.8 P/As mole ratio 7. 16 4.9 15 3.5 Min. Hill V1, V2 = 3312.4 Seed 0.5% into vessel-1. 18.5 (4 hours) (Test BS 7) V3 = 20Slurry feed rate 20.7 cc/min. recovery= 455 16.2 4 26.210.1 4 4 Solution feed rate 4.8 cc/min. 95.9% P/As mole ratio 7. Min. Hill 2 2 8-9 10 for 16 No seed, lime 0.76%. 8.2 4.9 2 11.8 (Test BS 8) Lime feed rate 16.0 cc/min for cc/min lime 503.7, slurry feed; first 2 hours, then 22 cc/min 506.8 19.2 4 4 7.8 4 7.7 12 for 22 for 2 hours. Solution feed rate cc/min 20 cc/min. P/As mole ratio 7. ASARCO No seed, lime 10.0%. 2 88.9 4 129.4 140.8 7.2 (4 hours) 4 (Test BS 10) Lime feed rate 20.0 cc/min. recovery= 12.4 8-10 2,188 ppm scrubber Solution feed rate 40 cc/min. $99.9996^{+}\%$ 4 162.0 blowdown P/As mole ratio 7. 1 114.2 3.0 (14 hours) 4 14.8 ASARCO No seed. lime 1.0%. recovery= 9.2 8 24.4 1 (Test BS 4) Lime feed rate 19.7 cc/min. 12.4 8,089 ppm 41 99.999996+% scrubber Solution feed rate 4.6 cc/min. 8 8.3 14 188.4 blowdown P/As mole ratio 7. 26.7 14

^{1.} Analytical data presented in Appendix B.

^{2.} Feed solution P/As ratio formulation error.

Table 7-2. Mineral-Like Precipitation Technology Applied to ASARCO Scrubber Blowdown Water: Final Effluent Concentrations

$Sample^{1}$	SP^2	Time	pН				Concen	tration (mg	g/L)			
		(hours)		As	Ca	Cd	Cu	Fe	Mn	P	Pb	Zn
MHA-128: Inlet	101	24	1.3	$\sim 3.3~\mathrm{gpl}$	195	215	1.3	35.6	1,786	48	4.1	69.3
MHA-129:P/As= ~ 5.5 : Effluent	106	24	12.1	$7\text{-}9^3\mu\text{g/L}$	776	20	NA	0.07	0.10	9	0.02	11.4
MHA-112A: Inlet	101	3	1.1	$\sim 3.3~\mathrm{gpl}$	203	230	1.3	39.2	1,884	389	8.0	73.9
MHA-113:P/As= ~ 11.9: Effluent	106	3	12.4	6-9 ⁴ μg/L	754	10	< 0.26	0.07	0.10	7.7	< 0.02	0.0

^{1.} P/As in the water entering the treatment system.

Table 7-3. Mineral-Like Precipitation Technology Applied to ASARCO Scrubber Blowdown Water: Summary of Solution Conditions

Time]	SP 101 Feed Wate	er	Ph	SP 102 nosphate Ad	ded		SP 103 Lime Tanl	(Re	SP 104 sidence Ta	nk	Re	SP 105 esidence T		Settl	SP 106 er Disch	arge
(hours)	pН	E_H , mV	T°C	pН	E _H , mV	T°C	pН	E_{H} , mV	TºC	pН	$E_{\rm H}$, mV	T°C	pН	E_H , mV	T°C		E_H , mV	0
1	1.5		26.0	1.2	590		12.2			12.2		33.0						
2	1.5		26.0	1.2	580		12.2				-95	33.0						
4	1.4		26.0	1.1	625	28.0	11.5		35.0	11.6		34.0						
8	1.5		28.0	1.1	610	28.0	11.5		35.0	11.8	-95	34.0						
12	1.5	630	24.0	1.2	625		11.4			11.4	-65	33.0	11.6			12.4	-95	31.0
24	1.3		26.0	1.0	615		11.8				-75	32.0						
]	RAISED P	HOSPHA	TE								
1	1.1		29.0	1.0		37.5	10.8	-55	37.0		-55	36.0						
2	1.0		29.0				12.2	-55	37.0		-70	36.0						
3	0.9	605	28.5				12.3	-65	37.0	12.3	-55	37.0		-55	36.0	12.4	-70	

Demonstration Test Conditions:

Water feed rate was 3.6-4.0 liters/min.

Phosphate feed rate was 70-75 cc/min (of 85% H₃PO₄) for treatment of 1665 gal scrubber blowdown water (SBW) (i.e., 27-hour test).

Phosphate feed rate was 150 cc/min (of 85% H₃PO₄) for treatment of 405 gal SBW (i.e., 3-hour test).

P/As mole ratio in the initial solution phase was nominally ~ 5.5 (varied between 5.0-6.0) for the 27-hour test.

P/As mole ratio in the initial solution phase was nominally ~ 11.9 (varied between 11.5-12.5) for the 3-hour test.

Lime addition rate was 2.4 L/min. of a 10 % lime slurry for the 27-hour test.

Lime addition rate was 4.0 L/min. of a 10 % lime slurry for the 3-hour test.

^{2.} SP= sampling port: 101 inlet sampling port, 106 effluent sampling port.

^{3.} Range for all 27-hour samples.

^{4.} Range for all 3-hour samples.

Table 7-4. Total Metals Concentration and TCLP Results for Product Solids from the Treatment of Scrubber Blowdown Water: P/As Mole Ratio~ 5.5

Sample	Description	As	Ba	Ca	Cd	Cr	Pb	Hg	Se	Ag
			Concen	tration [mg/kg	g (ppm)]					
MHA-121	Total Metals	41,300	2.56	384,000	-	15.9	273	-	321	< 1.69
			Concer	ntration [mg/L	(ppm)]					
MHA-121	TCLP	0.90	0.076	-	0.030	0.13	< 0.03	0.40	0.30	< 0.003
Reference	Maximum element	5	100	-	1	5	5	0.2	1	5

Percent solids in the sample supplied to MSE-HKM for TCLP test: $35.6 \pm 1.2\%$.

All TCLP tests were conducted by MSE-HKM in accordance with EPA ICP protocol.

Table 7-5. Total Metals Concentration and TCLP Results for Product Solids from the Treatment of Scrubber Blowdown Water: P/As Mole Ratio ~ 11.9

Sample	Description	As	Ba	Ca	Cd	Cr	Pb	Hg	Se	Ag
			Co	ncentration (m	g/kg)					
MHA-138	Total metals	21,300	1.59	307,000	-	7.4	110	-	179	< 1.7
			Co	oncentration (m	ng/L)					
MHA-138	TCLP	82.1	0.068	-	0.127	0.065	< 0.03	0.37	0.24	< 0.003
MHA-138	TCLP	87.5	0.210	-	0.147	0.062	< 0.03	0.49	0.20	< 0.003
Reference	Maximum element concentration	5	100	-	1	5	5	0.2	1	5

Percent solids in the sample supplied to MSE-HKM for TCLP test: $30.4 \pm 0.4\%$.

Composition of the solids: 2.7% As, 5.7% P, 23.0% Ca, 1.0% Mn, 0.09% Cd, 0.05% Zn, 0.01% Pb (Determined by fluorescence analysis by Ashe Analytics, Inc.)

All TCLP tests were conducted by MSE-HKM in accordance with EPA ICP protocol.

Table 7-6. Summary of Experimental Results for Long-Term Air Sparging of Ambient Temperature Precipitated Mineral-Like Products, ASARCO Scrubber Blowdown Water: Time=0

Commle	Tashnalagu Haadi					Elementa	I Concentra	tion, µg/L				
Sample	Technology Used ¹	pН	Al	As	Cd	Cu	Pb	Mn	Hg	P	Ag	Zn
MHA-121-1	MLP, $P/As = 5.5$	12.5	< 20	4	< 5	3	40	< 4	459	2,310	4	37
MHA-121-2	MLP, $P/As = 5.5$	12.6	< 20	3	< 5	5	40	< 4	436	2,250	< 3	41
MHA-121-3	MLP, $P/As = 5.5$	12.7	< 20	4	< 5	2	50	4	471	2,460	< 3	44
MHA-139-1	MLP, $P/As = 11.9$	12.4	20	2	< 5	5	< 30	< 4	295	2,200	5	< 13
MHA-139-2	MLP, $P/As = 11.9$	12.4	30	4	< 5	2	< 30	< 4	283	2,720	4	< 13
MHA-139-3	MLP, $P/As = 11.9$	12.4	< 20	3	< 5	3	< 30	< 4	265	1,190	< 3	< 13

^{1.} MLP= mineral-like precipitation. P/As mole ratio in the initial solution phase was nominally ~ 5.5 (varied between 5.0-6.0) for the 27-hour test; P/As mole ratio in the initial solution phase was nominally ~ 11.9 (varied between 11.5-12.5) for the 3-hour test. The solids placed under long-term aging were formed from the waters containing the different P/As ratios.

Table 7-7. Mineral-Like Precipitation Technology Applied to ASARCO Thickener Overflow Water: Summary of Arsenic Removal

				[As], μg/	L		
Time (hours)	Treated gallons ¹	SP 101	SP 102	SP 103	SP 104	SP 105	SP 106
Time (nours)	Treated ganons	Feed Water	Phosphate Added	Lime Tank	Residence Tank	Residence Tank	Settler Discharge
1	285	85	320	250	84	93	210
8	705	220		28	7	17	2
8	705						11
16	1,185	235	100			6	6
16	1,185						8
16	1,185						8
16	1,185						15
16	1,185						7
Average residence time, minutes	3			66	66	66	390
		RAISED	PHOSPHATE				
1	285	200	540	See Table 7-13 ²		4	4
12	945	140	420				4

				[As], μg/	L		
Time (hours)	Treated gallons ¹	SP 101	SP 102	SP 103	SP 104	SP 105	SP 106
Time (nours)	Treated gallons	Feed Water	Phosphate Added	Lime Tank	Residence Tank	Residence Tank	Settler Discharge
12	945		347				4
20	1,425	380	540			4	13
20	1,425		570			4	4
20	1,425					4	3
20	1,425					4	3
20	1,425						12
20	1,425						12
Average residence time, minutes				65	65	65	380

- 1. Time zero taken to be after one volume displacement of water added, i.e., all tanks full; 225 gallons.
- 2. The required residence time for removal of arsenic from solution is less than 15 minutes, see Table 7-13.

Demonstration Test Conditions:

Water feed rate was 3.6-4.0 liters/min.

Phosphate feed rate was 200 cc/min (of 300 cc H₃PO₄/80 gallons deionized water) for treatment of first 1,185 gallons of wastewater.

Phosphate feed rate was 130 cc/min (of 3 liters H₃PO₄/80 gallons deionized water) for treatment of 1,425 gallons of wastewater.

Total P in the inlet water was ~ 5.8 mg/L

Total P/As mole ratio was 10 (ratio in solution phase was \sim 34) for first 1,185 gallons of wastewater.

Total P/As mole ratio was 100 (ratio in solution phase was ~ 615) for last 1,425 gallons of wastewater.

Lime addition rate was 365 cc/min of a 1% lime slurry for treatment of 2,600 gallons of wastewater.

Table 7-8. Mineral-Like Precipitation Technology Applied to ASARCO Thickener Overflow Water: Final Effluent Concentrations

Sample ¹	SP^2	Time	pН			Cor	centratio	n, μg/L				
		(hours)		As	Ca	Cd	Cu	Fe	Mn	P	Pb	Zn
MLA-209A: Inlet	101	20	11.5	~5.8 mg/L 3	732 mg/L	20	10	30	20	25 mg/L	< 20	< 9
MLA-210: P/As= 100: Effluent	106	20	12.1	$3-13^{4}$	813 mg/L	< 4	10	50	10	4 mg/L	< 20	< 9

- 1. P/As in the water entering the treatment system.
- 2. SP= sampling port: 101 inlet sampling port, 106 effluent sampling port.
- 3. Total arsenic= 5.9 mg/L; Dissolved arsenic= 0.26 mg/L.
- 4. Range for all the 20 hr samples.

Table 7-9. Mineral-Like Precipitation Technology Applied to ASARCO Thickener Overflow Water: Summary of Solution Conditions

Time]	SP 101 Feed Water		SP 102 SP 103 Phosphate Added Lime Tank						F	SP 104 Residence Ta	ank	SP 105 Residence Tank			SP 106 Settler Discharge		
(hours)	pН	$E_{\text{H}}\text{, }mV$	Т⁰С	pН	$E_{\text{H}}\text{, }mV$	T°C	pН	$E_{\scriptscriptstyle H}\text{, }mV$	Т℃	pН	$E_{\text{H}}\text{, }mV$	TºC	pН	E_{H} , m V	Т℃	pH E _H ,mV	Т℃	
1	11.6	-55	24.5	10.5	-5		12.4				-85	24.0						
2	11.6	-60	24.5	10.2	-5		12.4				-75	24.0						
4	12.5	10	24.0	10.2	15		12.3				-60	23.5						

Time (hours)	SP 101 Feed Water			SP 102 Phosphate Added		SP 103 Lime Tank			SP 104 Residence Tank			SP 105 Residence Tank			SP 106 Settler Discharge			
	pН	$E_{\text{H}}\text{, }mV$	Т°С	pН	$E_{\text{H}}\text{, }mV$	Т℃	pН	$E_{\text{H}}\text{, }mV$	T°C	pН	$E_{\scriptscriptstyle H}\text{, }mV$	T℃	pН	E_{H} , m V	T°C	pН	$E_{\scriptscriptstyle H}\text{,}mV$	Т°С
8	11.5	20	23.0	10.3	65		12.5			12.5	-20	23.0				11.5	10	22.0
12	11.6	-5		10.5	20		12.5				-50	21.0				12.4		
16	11.4		24.0	10.4	70		11.0				20	22.0				12.2		22.0
RAISED PHOSPHATE																		
1	9.0	-10	25.0	5.5	130.0		11.8	-45		11.8	-60							
2	11.1	-10	24.5	5.4	115.0		11.8	-40		11.8	-50	25.0						
4	11.3	-10	25.0	5.5	140.0		11.8	-45		11.8	-55	24.5						
8	11.1	40	23.0	5.5	165.0		11.9	-40		11.9	-45	23.0						
12	9.8		23.0	5.5	175.0		11.8				35	23.0						
16	11.1		25.0	5.5	130.0		11.8			11.8	-30	24.0				11.8	-20	24.0
20	11.0	(10)	24.5				11.7	-95		11.7	-90	24.5				11.7	-70	24.5

Solids in settler bottom at end of the second test was 6.2%

Demonstration Test Conditions:

Water feed rate was 3.6-4.0 liters/min.

Phosphate feed rate was 200 cc/min (of 300 cc $H_3PO_4/80$ gallons deionized water) for the first 16-hour test.

Phosphate feed rate was 130 cc/min (of 3 liters H₃PO₄/80 gallons deionized water) for the 20-hour test.

Total P in the inlet water was ~ 5.8 mg/L

Total P/As mole ratio was ~ 10 (ratio in solution phase was ~ 34) for the first 16-hour test.

Total P/As mole ratio was ~ 100 (ratio in solution phase was ~ 615) for the 20-hour test.

Lime addition rate was 365 cc/min. of a 1% lime slurry for both the 16 and 20-hour tests.

Table 7-10. Total Metals Concentration and TCLP Results for Product Solids from the Treatment of ASARCO Thickener Overflow Water

Sample	Description	As	Ba	Ca	Cd	Cr	Pb	Hg	Se	Ag
		C	oncentration	(mg/kg)						
MLA-218	Total metals	370	16.8	343,000	-	< 5.7	58.9	-	384	< 1.91
		C	oncentration	(mg/L)						
MLA-218	TCLP	3.87	0.100	-	< 0.005	0.019	< 0.03	0.008	0.49	< 0.006
Reference	Maximum element concentration	5	100	-	1	5	5	0.2	1	5

Sample Description As Ba Ca Cd Cr Pb Hg Se Ag

Percent solids in the sample supplied to MSE-HKM: $29.3 \pm 0.6\%$.

Composition of the solids: 0.033% As, 5.4% P, 23.0% Ca (Determined by fluorescence analysis by ASHE Analytics) All TCLP tests were conducted by MSE-HKM in accordance with EPA ICP protocol.

Table 7-11. Summary of Experimental Results for Long-Term Air Sparging of Ambient Temperature Precipitated Mineral-Like Products, ASARCO Thickener Overflow Water: Time=0

Sample	Technology Used	ъШ	Elemental Concentration, µg/L										
Sample	reciniology Osed	pН	Al	As	Cd	Cu	Pb	Mn	Hg	P	Ag	Zn	
MHA-217-1	Mineral-like precipitation	10.5	< 20	4	< 5	6	< 30	< 4	< 0.1	1,660	< 3	< 13	
MHA-217-2	Mineral-like precipitation	10.6	< 20	7	< 5	5	< 30	< 4	< 0.1	1,600	< 3	< 13	
MHA-217-3	Mineral-like precipitation	10.6	< 20	4	< 5	4	< 30	< 4	< 0.1	1,560	< 3	< 13	

Table 7-12. Mineral-Like Precipitation Technology Applied to Mineral Hill Mine 1,300' Portal Groundwater: Summary of Arsenic Removal

,		[As], μg/L									
			SP 102	SP 103	SP 104	SP 105	SP 106				
Time, hrs	Treated gallons¹	Feed Water	Phosphate Added	Lime Tank	Residence Tank	Residence Tank	Settler Discharge				
1	285	470	470		5	6	6				
1	285						6				
1^2	285	443	485			27	8				
8	705	470				8	5				
8	705						< 1				
8	705						3				
8	705						8				
16	1,185	450	480			4	22				
			RAISED P	HOSPHATE							
24	1,665	470	440	7	4	3	4				
24^2	1,665	448	411		318,319	< 4	< 4				
32	2,145	460					4				
32^{2}	2,145	432					< 4				
LOWERED LIME TO 0.5 g/L											
40	2,625	460				6	4				
40	2,625	460					< 1				
40^{2}	2,625	398	411		< 4	< 4	< 4				
48	3,105	460					5				
48	3,105	420	460	11			6				
48	3,105	460				7	7				
48^{2}	3,105	445	417,466	18		11	9				
56	3,585	490				7	6				
64	4,065	480					6				
64^{2}	4,065	426					11				
		LO	OWERED LI	ME TO 0.25	g/L						
68	4,305	450	440		9		25				
68	4,305						4				
68	4,305						7				
68	4,305						4				
68^2	4,305	451	429495		10		5, 3				
76	4,785	480				8	4				
76^2	4,785	453,506				10	8				

Table 7-12. Mineral-Like Precipitation Technology Applied to Mineral Hill Mine 1,300' Portal Groundwater: Summary of Arsenic Removal

				[A	As], μg/L		
		SP 101	SP 102	SP 103	SP 104	SP 105	SP 106
Time, hrs	Treated gallons ¹	Feed Water	Phosphate Added	Lime Tank	Residence Tank	Residence Tank	Settler Discharge
84	5,100	490	450			13	2
84	5,100						< 1
84	5,100						2
84^{2}	5,100		438436			13	4
Average residence	e time, minutes	s		65	65	65	380

- 1. Time zero taken to be after one volume displacement of water added, i.e., all tanks full; 225 gallons.
- 2. Analyses performed at Montana Tech

Demonstration Test Conditions:

Water feedrate was 3.6-3.8 liters/min.

Phosphate feed rate was 70 cc/min (of 1 cc H₃PO₄/gallon deionized water) for treatment of first 1,185 gallons of wastewater.

Phosphate feed rate was 70 cc/min (of 2 cc H_3PO_4 /gallon deionized water) for treatment of 3,915 gallons of wastewater.

Total P/As mole ratio was approximately 10.6-12.9 for first 1,185 gallons of wastewater.

Total P/As mole ratio was approximately 21.2-36.4 for last 3,915 gallons of wastewater.

Lime addition rate (1 g/L)was 360-380 cc/minute of a 1% lime slurry for treatment of 2,145 gallons of wastewater.

Lime addition rate (0.5~g/L) was 180-190 cc/minute of a 1% lime slurry for treatment of 1,920 gallons of wastewater.

Lime addition rate (0.25 g/L) was 90-100 cc/minute of a 1% lime slurry for treatment of 795 gallons of wastewater

Table 7-13. Mineral-Like Precipitation Technology Applied to Mineral Hill Mine 1,300' Portal Groundwater Final Effluent Concentrations

Sample ¹	SP^2	Time,	pН				Conce	entration,	μg/L			
		Hrs		As	Ca	Cd	Cu	Fe	Mn	P	Pb	Zn
MLM-328: Inlet	101	48	8.2	420	125 mg/L	10	10	< 24	20	< 30	< 20	10
MLM-329: P/As= 10- 20:Effluent	106	48	12.0	6-73	314 mg/L	< 39	< 24	40	10	500	< 20	10
MLM-344: Inlet	101	68	7.6	450	119 mg/L	< 4	< DL	< 24	NA	380	< 20	40
MLM-346: P/As= 20: Effluent	106	68	12.0	4-74	215 mg/L	40	< 26	< 242	NA	< 310	< 206	< 88

- 1. P/As in the water entering the treatment system.
- 2. SP= sampling port: 101 inlet sampling port, 106 effluent sampling port.
- 3. Range for all 48 hour samples.
- 4. Range for all 68 hour samples.

Table 7-14. Mineral-Like Precipitation Technology Applied to Mineral Hill Mine 1,300' Portal Groundwater: Summary of Solution Conditions

Time		SP 101 eed Wa			SP 102 phate A			SP 103 me Ta		Res	SP 104 idence			SP 105 idence		Sett	SP 106 ler Disc	
(hrs)		$\begin{array}{c} E_{H}, m \\ V \end{array}$		pН	E _H , mV		pН	_			$\begin{matrix} E_H, m \\ V \end{matrix}$						$\begin{array}{c} E_H, m \\ V \end{array}$	
1	8.2		18.0	8.0	230	18.0	12.6				25	18.0						
2	8.2		17.5	7.8	30		12.7				-35	18.0						
4	8.2		16.5	7.7	115		12.8				-15	18.0						
8	8.2		17.0	7.7	110		12.5				-10	17.0						
12	8.2		15.0	7.8	200		12.5				-50							
16	7.6		17.0				12.4	60			25	16.0						
]	RAISE	D PHO	OSPH/	ATE (S	See not	e)						
24	7.3		18.0				12.4	40			25	18.0						
32	8.2		15.0	7.4	175.0		12.4			12.2	25	15.0	12.4	25	16			
							LOW	ERED	LIMI	Е ТО	0.5 g/L							
40	7.5		15.0				12.0	35			25	14.8						
48	7.4		17.0				11.9, 12.1	3545		12.2	2535	16.0	12.3	25				
56	8.2		15.5	7.5	330.0		12.0					15.5						
64	7.4		16.5				12.1	80			55	15.5						
							LOW	ERED	LIME	ТО	0.25 g/l	Ĺ						
69	7.4						11.5			11.8			11.9					
72	8.1	185	18.5	7.2	220.0		11.3			11.4	120	18.5	11.6			11.9		
82	8.1			7.2			11.2			11.1			11.1			11.7		
84	8.1		17.0	7.6	250.0		11.1				150	16.5						

Demonstration Test Conditions:

Water feedrate was 3.6-3.8 liters/min.

Phosphate feed rate was 70 cc/min (of 1 cc H_3PO_4 /gallon deionized water) for the first 16 hours of the test.

Phosphate feed rate was 70 cc/min (of 2 cc H₃PO₄/gallon deionized water) for the reminder of the test.

Total P/As mole ratio was approximately 10.6-12.9 for first 16 hours of the test.

Total P/As mole ratio was approximately 21.2-36.4 for the reminder of the test.

Lime addition rate (1 g/L) was 360-380 cc/minute of a 1% lime slurry for the first 32 hours of the test.

Lime addition rate (0.5 g/L) was 180-190 cc/minute of a 1% lime slurry for the second 32 hours of the test.

Lime addition rate (0.25 g/L) was 90-100 cc/minute of a 1% lime slurry for the reminder of the test.

Table 7-15. Arsenic Concentration as a Function of P/As Mole Ratio

Sample Designation	Time, min	[As], μg/L					
Sample Designation	Time, iiiii	P/As = 10	P/As = 20	P/As = 200			
MLMMT-83	0	562	553	546			
MLMMT-84	5	29	9	8			
MLMMT-85	15	12	12	6			
MLMMT-86	30	11	12	4, 5, 7			
MLMMT-87	60	24	10	5			

Hydrated lime concentration was constant in each test at 0.5 g/L

Table 7-16. Arsenic Concentration as a Function of Hydrated Lime Content

Sample Designation	Time min	[As], μg/L		
Sample Designation	Time, min	$Ca(OH)_2 = 0.1 \text{ g/L}$	$Ca(OH)_2 = 0.5 g/L$	$Ca(OH)_2 = 1.0 g/L$
MLMMT-83	0	504	472	441, 502, 508
MLMMT-84	5	372	6	6
MLMMT-85	15	400	5	7
MLMMT-86	30	364	6	12
MLMMT-87	60	379	5, 10, 2	17

The P/As mole ratio was constant in each test at 10.

Table 7-17. Total Metals Concentration and TCLP Results for Product Solids from the Treatment of Mineral Hill Mine 1,300' Portal Groundwater

Sample	Description	As	Ba	Ca	Cd	Cr	Pb	Hg	Se	Ag
				Concentrat	ion, mg/kg					
MLM-354A	Total Metals	728	18.5	273,000	-	7.7	16.4	-	< 15.8	< 1.2
				Concentration	on, mg/liter					
MLM-354A	TCLP	0.15	0.06	-	< 0.005	0.03	< 0.03	< 0.0001	< 0.05	< 0.003
Reference	Max Element Conc.	5	100	-	1	5	5	0.2	1	5

Composition of the solids: 0.075% As, 0.6% P, 15.5% Ca (Determined by fluorescence analysis by Ashe Analytics, Inc.) All TCLP tests were conducted by MSE-HKM in accordance with EPA ICP protocol.

Table 7-18. Summary of Experimental Results for Long-Term Air Sparging of Ambient Temperature Precipitated Mineral-Like Products, ASARCO Thickener Overflow Water: Time= 0

Sample	Technology	ъU	Elemental Concentration, μg/L									
Sample	Used	рп	Al	As	Cd	Cu	Pb	Mn	Hg	P	Ag	Zn
MLM-354-1	Mineral-Like Precipitation	12.2	5	4	< 5	21	< 30	< 4	0.3	30	< 3	< 13
MLM-354-2	Mineral-Like Precipitation	12.1	7	7	< 5	17	< 30	< 4	< 0.1	< 30	< 3	< 13
MLM-354-2	Mineral-Like Precipitation	12.2	2	4	< 5	18	< 30	< 4	< 0.1	40	< 3	< 13
ILM-136A-1	Ferrihydrite Adsorption	9.1	40	230	< 5	16	40	< 4	< 0.1	50	< 3	< 13
ILM-136A-1	Ferrihydrite Adsorption	9.1	20	230	< 5	16	< 30	< 4	0.1	40	< 3	< 13
ILM-136A-1	Ferrihydrite Adsorption	9.2	20	230	< 5	14	< 30	< 4	< 0.1	40	< 3	< 13

Table 7-19. Analytical Results for ASARCO Thickener Overflow Water Demonstration Using Low Iron Ratio

Time (Minutes)	Arsenic Concentration (Fg/L)
0	6,300
5	100
12	200
40	100
60	100
240	300
360	600
480	500
600	200
960	400
1,440	300

Table 7-20. Analytical Results for ASARCO Thickener Overflow Water Demonstration Using High Iron Ratio

Time (Minutes)	Arsenic Concentration (Fg/L)
0	6,300
5	10
12	5
40	20
60	46
240	340
360	260
480	210
600	150
960	140
1,440	201

Table 7-21. Analytical Results from Mineral Hill Mine Using Ferrihydrite Adsorption

Time (Minutes)	Arsenic Concentration (Fg/L)
0	600
1,100	40
1,590	55
1,860	52
2,550	46
3,030	73

8. Economic Analysis

One objective of this study was to perform a first order cost estimate for each of the treatment flow sheets. A "first order" cost estimate was performed using the flow sheet presented in Figures 3-1 through 3-4. The cost estimate presented here is not a detailed engineering cost analysis. It is a first order cost estimate that should be within \pm 30%.

Definitions and cost estimation factors are taken primarily from "Mineral Processing Equipment Cost and Preliminary Capital Cost Estimation" (Ref. 12). Itemized equipment lists were used where possible and literature quoted cost figures were used where available. All costs were updated to the third quarter 1997 using the Marshall and Swift (M&S) Index value of 1059.6 (Ref. 13).

Major cost items have been included. The factored capital cost (FCC) totals include minor equipment, instrumentation, processing piping, auxiliary engineering, and plant size factors. An example is presented in Table 8-1.

Capital costs (using Table 8-1) and operating costs were estimated. Equipment costs were based on cost equations of the form:

$$Cost_{now} = a(capacity)^b(M&S_{now}/M&S_{then})$$

Where, a and b are constants for a particular piece of equipment (taken from Ref. 12).

Assumptions made for the cost estimate are presented in Table 8-2.

8.1 Factored Capital Cost

An equipment list was prepared for each unit operation, and the FCC cost was estimated as

described above. The FCC was determined by using the factors as presented in Table 8–1 (selected factors for this study are highlighted).

8.2 Operating Cost

Annual operating cost estimates were established based on reagent consumption, manpower requirements, maintenance and power consumption. Reagent consumption was based on calculated mass flow. Reagent costs were taken from the Chemical Market Reporter (Ref. 14). Manpower requirements, maintenance, and power consumption were estimated using the following factors (i.e., manpower 25% FCC, maintenance 5% FCC, and power 4% FCC).

8.3 Net Present Value

The net present value (NPV) was determined by the relationship:

NPV= FCC + USPW Operating Cost, where USPW= Uniform Series Present Worth

USPW= $[(1+I)^n-1/I(1+I)^n]$ I= interest rate, n= number of years

Assumptions: Cost Estimate Assumptions are presented in Table 8-2.

8.4 Results

The three different technologies, mineral-like precipitation, alumina adsorption with microfiltration, and ferrihydrite adsorption were economically evaluated for a system which contained 0.5 ppb arsenic at a flow rate of 300 gallons per minute. The comparative results can be seen in Table 8-3.

Table 8-1. Factored Capital Cost Estimate Form 1. Purchased equipment costs 2. Installed equipment costs Item 1 multiplied by 1.43 3. Process piping Type plant: Percent of Item 2: Solid 7%-10% Solid-Fluid 10%-30% Fluid 30%-60% 4. Instrumentation Amount of automatic control: Percent of Item 2: None 2%-5% Some 5%-10% Extensive 10%-15% 5. Buildings and site development Type plant: Percent of Item 2: Outdoor 5%-20% Outdoor-Indoor 20%-60% Indoor 60%-100% 6. Auxiliaries (e.g., electric power) Extent: Percent of Item 2: Existing 0% Minor additions 0%-5% Major additions 5%-25% New facilities 25%-100% 7. Outside lines Average length: Percent of Item 2: Short 0%-5% Intermediate 5%-15% Long 15%-25% 8. Total physical plant costs: Sum of Items 2+3+4+5+6+79. Engineering and construction Complexity: Percent of Item 8: Simple 20%-35% Difficult 35%-60% 10. Contingencies Type process: Percent of Item 8: Firm 20% Subject to change 20%-30% Speculative 30%-50% Average 30% 11. Size factor Size plant: Percent of Item 8: Large commercial 0%-5% Small commercial 5%-15% Pilot plant 15%-35%

12. Factored Capital Costs (FCC): Sum of Items 8+ 9+ 10+ 11 Note: Percentages selected for this study are highlighted.

Table 8-2. Cost Estimate Assumptions

Cost, \$

Item Assumptions

Site Treatment will be conducted at a currently operating facility. Major buildings (containing

sufficient space for the treatment process) are available. Analytical capabilities exit.

Tailings ponding facilities are in place.

Permitting Regulatory permits are in place.

Flow Mineral Hill Mine water a: 300 gal/min, 330 days/yr, containing 0.5 ppb arsenic.

Solution P/As mole ratio = 10

Cost Interest rate= 10%

Life of system= 10 years

NPV= FCC + USPW Operating Cost

Operating Cost factors:

Reagents determined from mass flow.

Manpower= 25% FCC Maintenance= 5% FCC Power= 4% FCC

Not considered: depreciation, leases, salvage, tax

FCC: Factored Capital Cost NPV: Net Present Value

USPW: Uniform Series Present Worth

Table 8-3. Economic Evaluation for Selected Technologies Treating Groundwater with 0.5 ppb Arsenic at 300 gal/min

	Mineral-Like Precipitation	Alumina Adsorption	Ferrihydrite Adsorption
Capital	$250,000 \pm 75,000$	$$396,000 \pm $118,800$	\$250,000± 75,000
Operations and Maintenance per Year	\$41,080	\$130,700	\$78,904
Operations and Maintenance per 1,000 gallons treated	\$0.30 +/- 0.09	\$0.70 +/- 0.30	\$0.55 + /- 0.16

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APPENDIX A

Arsenic Removal Project Demonstration Sampling, Analytical, And Calibration Procedures

1. ARSENIC REMOVAL PROJECT DEMONSTRATION SAMPLING, ANALYTICAL, AND CALIBRATION PROCEDURES

1.1 SITE SELECTION AND SAMPLING PROCEDURES

Two demonstration sites have been chosen for this project: the ASARCO Lead Smelter in East Helena, Montana and the Mineral Hill Mine located near Gardiner, Montana. Several sampling locations have also been identified at different points in the treatment trains for the three technologies being demonstrated. Selection of the correct sampling sites is necessary to ensure that the project objectives are met. Sampling sites must be selected to meet the following general criteria:

- chance of external contamination should be minimized;
- location should be representative of the entire waste stream;
- location should be as close as possible to the treatment process or the component in the treatment train being monitored to prevent further chemical changes from occurring in the waste stream; and
- sampling sites should be chosen so that the effect of each component in the treatment train can be analyzed.

1.2 SAMPLING PROCEDURES AT THE DEMONSTRATION SITE

Sampling preservative, containers, method types and references are summarized in Table A-1. The approximate sampling locations for the field demonstrations are shown in Figures A-1, A-2, and A-3. The sampling locations shall not change unless authorized by the Project Manager, with appropriate documentation that justifies the change. Tables A-4, A-5, A-6, A-7, A-8, A-9, and A-10 show sampling frequency, by position and sample type.

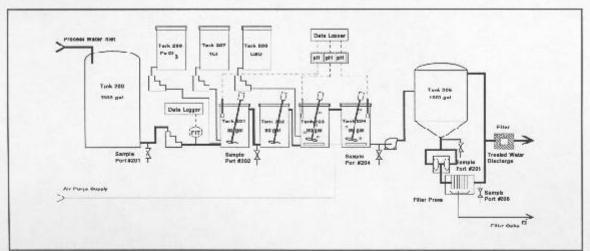


Figure A-1. Ferrihydrite process flow diagram.

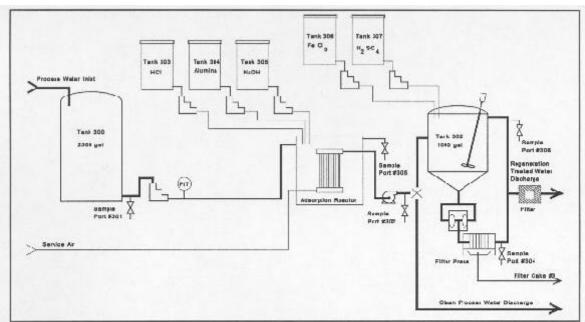


Figure A-2. Alumina absorption process flow diagram.

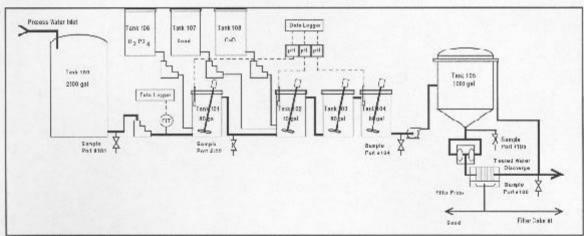


Figure A-3. Mineral-like precipitation process flow diagram.

1.2.1 Sampling Procedures

All solid and aqueous samples obtained during the demonstration will conform to the appropriate MSE Standard Operating Procedures (SOP) listed below (Table A-2) and found in Appendix C of the QAPP. All sample containers for critical measurements will be high-density polyethylene (HDPE) bottles. These bottles will only be used once (unless otherwise indicated) to prevent cross contamination of samples.

All procedures outlined in the SOP's shall be followed during sampling activities. Each aqueous sample will be taken as a single grab sample. Solid samples will be taken from homogenized solid material generated at ASARCO and the Mineral Hill Mine during each filter press operation.

Aqueous samples will be obtained at several locations in the processes depicted in Figures A-1, A-2, and A-3. The sampling tap at the sampling location will be flushed (allowed to flow briefly) before the samples are collected. Tap sampling procedures specified in EPA Method S004 Sampling and Analysis Methods for Hazardous Waste Combustion, will be followed. Sample containers will be triple rinsed with small aliquots of sample, and then the sampling container will then be filled from the center of flow.

Table A-1. Preservatives, holding times, containers, method types and references.

Parameter	Matrix	Preservative	Holding Time	Sample Size & Container	Method Type	Reference
Arsenic Speciation	Aqueous	#4EC, Filter, pH#2 HCl	Analyze immediately	250-mi_ HDPE	Ion Exchange, Furnace AA	Ficklin Ion Exchange and Appendix C
Iror. Speciation	Aqueous	#4EC, Filter, pH#2 HCl	Analyze immediately	250-mL HDPE	Colorimetric	Standard Methods 3500- Fe D, Appendix C
TSS	Aqueous	#4°C	7 days	500 mL HDPE	Gravimetrie	EPA Method 160.2
Sulfate	Aqueous	#4°C	28 days	500-mL HDPE	Colorimetric	EPA Method 375.2
pH	Aqueous	None	Analyze immediately	2000-mL HDPE	pH meter	EPA (SW-846) Method 9040
En	Aqueous	None	Analyze immediately	100-mL HDPE	Eн meter	Equip. Manufacturer instructions
Flowrate	N/A	None	Analyze immediately	N/A	Flow meter	Manufacturer's Instructions
Total Recoverable Metals (Al, As, Cd, Cu, Fe, Pb, F, Zn by ICP)	Aqueous	#4EC, pH#2 HNO ₃	6 morths	500-mL HDPE	ICP	EPA SW-846 Preparation Method 3005 ¹ /ICP Method 6010
Dissolved Metals (As hy AA)	Aqueous	#4EC, Filter, pH#2 HNOs	6 morths	500-mL HDPE	Furnace AA	EPA SW-846 Preparation/ AA Method 7060 ²
Dissolved Metals (Al, As, Cd, Cu, Fe, Po, P, Zn by ICP)	Aqueous	#4EC, Filter, pH#2 HNO ₃	6 morths	500-mL HDPE	ICP	EPA SW-846 Preparation Method 3005 ¹ /ICP Method 6010
Total Meta.s (Al, As, Ea, Cd, Cr, Cu, Fe, P, Ph, Se, Ag, Zn, Ca by ICP)	Solid	None	6 morths	8 oz CWM	ICP	EPA SW-846 Preparation Method 3050/ICP Method 6010
X-ray Diffraction	Solid	#4°C	6 morths	20-mL HDPE	X-ray Diffraction	Appendix B
Stability Tests	Solid	None	6 morths	200 g 16 oz CWM	Montana Tech Procedure	Appendix B
% Solids	Səlid	None	6 months	Taken from solid sample	Drying/ Weighing	CLP SGW 3/90 Exhibit D, Part F and Appendix G
TCLP Metals	Solid	None	7 days to ext. 40 days after	At least 100 g 16 cz CWM	ICP	EPA SW-846 Extraction Method 1311/P-eparation Method 3005 ³ /ICP Method 6010

¹ Digestion method will be modified to result in a digestate concentration of 1% nitric acid, rather than 2% nitric acid.

⁴ Matrix modifier for arsenic will be added at instrument, so the digestion procedure will be modified by bring the sample back to the original volume of 100 mL following digestion.

Digestion method will be modified by the addition of 10-20 mL 30% H2Oz.

Table A-2. Summary SOPs.

SOP Number	Subject
C-1	Field Logbook/Photographs
C-2	Sample Packaging & Shipping
C-3	Field Quality Control Samples
C-4	Sample Custody

Sludge samples will be obtained from Filter Cake Locations 1, 2, and 3 following the precipitation and the separation from the treated effluent. Solid samples will be obtained from filter cakes that remain after the sludge has been dewatered by high pressure filtration. Solid samples from each filtered batch will be collected for TCLP, total metals, percent solids, and stability testing. Each solid or liquid sample that is collected will be given a unique sample ID that will distinguish it from all of the other samples collected for the project. Refer to section 4.5 for a discussion of sample labeling and sample identification numbers.

After the samples have been collected they will be preserved as shown in Table A-1. All aqueous samples shall leave no head space in the container to minimize air entrainment. Entrained air could react with species in the samples and affect the analytical results. The general procedure for obtaining aqueous samples for critical measurements from the demonstration will be as follows:

Dissolved As:

- 1. Obtain a clean 500 mL HDPE sampling container.
- 2. Open the sampling tap and flush the tap thoroughly.
- Collect a small amount of sample in the sampling container, cap, shake to rinse, and discard the rinsate. Repeat the rinsing procedure two more times.
- 4. Fill container at the prescribed sampling location.
- 5. Obtain another clean 500-mL HDPE sampling container.
- 6. Using a 0.45-µm filtering apparatus, filter at least 100 mL of water. Remove side arm and pour the filtrate directly into the sampling container. Repeat until 500 mL of sample has been filtered. Do not reuse the filter at another sampling location. Fill container completely so no headspace will remain when capped.
- 7. Adjust the pH of the filtered sample to <2 using nitric acid (HNO₃).
- 8. Cap bottle and seal per SOP for shipment to MSE Laboratory.
- 9. Log sample number, location, date, time, preservative, etc. per MSE SOP.
- Repeat procedure for each duplicate or sampling location. The 500-mL HDPE comainer may be used for each sample if washed between samples as per SOP.
- Store the samples in the sample refrigerator at the demonstration site at 4 °C until ready for shipment to the MSE HKM Laboratory.

The general procedure for collecting aqueous samples during the demonstration will be as follows:

- 1. Obtain a clean sampling container of the appropriate size.
- 2. Open the sampling tap and flush the tap thoroughly.
- Collect a small amount of sample in the sampling container, cap, shake to rinse, and discard the rinsate. Repeat the rinsing procedure two more times.
- 4. Fill the container at the prescribed sampling tap.

- 5. Filter the sample if necessary. Using a 0.45 µm filter if required (see Table A-1).
- Preserve the sample as appropriate (see Table A-1).
- 7. Cap the bottle, label bottle, and seal per SOP for shipment to MSE-HKM Laboratory.
- 8. Log sample number, location, date, time, preservative, etc. per MSE SOP.
- Store the collected samples in the sample refrigerator at the demonstration site until ready for shipment to MSE-HKM Laboratory.

The general procedure for collecting solid samples including critical TCLP samples during the demonstration will be as follows:

- 1. Combine the filter cakes generated from each filtering episode and homogenize by mixing.
- 2. Obtain a clean 16-ounce (oz) wide mouth sampling container.
- Scoop the solid material into the container until sufficient sample is collected (approximately 650 g). If efficient sample is not available, collect as much of the solid material as possible and record the reason sufficient sample was not collected in the logbook and on the chain of custody.
- 4. Cap container and seal per SOP for shipment to MSE Laboratory.
- 5. Log sample number, location, date, time, preservative, etc. per MSE SOP.
- Repeat procedure for each duplicate or sampling location.
- Store collected samples in the sample refrigerator at the demonstration site at 4 °C until ready for shipment to MSE-HKM Laboratory.

All samples will be shipped by project personnel via ground transportation in scaled coolers containing blue ice. The samples collected from the portion of the demonstration at Mineral Hill Mine can be driven to the MSE-HKM Laboratory in a matter of hours; samples collected from the portion of the demonstration at the ASARCO lead smelting plant can be driven to the MSE-HKM Laboratory in 90 minutes. Samples that have prohibitive holding times (arsenic and iron speciation) may be shipped by an overnight carrier if project personnel are not available to transport the samples.

1.2.2 Stability Test Sampling

For the stability tests being performed at Montana Tech, pH, En, and temperature will be monitored in the reaction vessel. After one year, the dissolved arsenic samples will be collected using a syringe and filtered using a 0.2- μ m Teflon syringe filter. After the testing period, the slurry will be filtered using suction filtration. The sampling procedure for recovering the solids for x-ray diffraction analysis and the final dissolved arsenic sample are described below:

- 1. Pour the slurry from the stability tests into the filtering funnel.
- When the entire sample has been filtered, close the valve on the hood to release the suction and separate the filter flask from the filtering funnel. Set filtrate aside for dissolved arsenic sampling and analysis.
- 3. Remove the filter from the filtering funnel with forceps.
- 4. Place the filter on a clean watch glass in a 105 °C even for at least 1 hour.
- Remove the dried sample from the oven and place the dried sample and filter paper in the vacuum desiccator for storage, cooling, and further drying.
- 6. Scrape as much of the solid material as possible from the filter into an tared HDPE container.
- Weight die sample and record the weight of the sample and container in the project logbook.

- 3. Label the samples with information summarized in Section 4.1.3 of the QAPP.
- Store the sample in the refrigerator in the Montana Tech Metallurgy Department until ready for x ray diffraction analysis.

Aqueous samples will be collected as syringe filtrates from either the reaction vessel after one year or the filtering flask after the 2 year aging period. Contents of the reaction vessel will be homogenized by the air that is continuously sparging in. Contents of the filtering flask will be manually homogenized by shaking prior to sampling. Syringe filter sampling is described below:

- 1. Obtain a clean 20-mL sampling vial, syringe, plunger, and 0.2-μm Teflon membrane syringe filter.
- Process approximately 20 mL of cistilled/deionized (DI/DS) water through the syringe and syringe filter.
- Homogenize the sample.
- 4. Process approximately 5 mL of sample through the syringe filter apparatus into the sample vial.
- Cap the sample vial and shake to rinse the vial with the sample. Open the sample container and discard rinsate.
- Homogenize the sample.
- Process enough homogenized sample through the syringe filter to fill the sample vial, leaving no headspace. Preserve the sample if necessary. Label the sample as outlined in Section 4.1.3 of the OAPP.
- Store the sample in the refrigerator in the Metallurgy Department at Montana Tech until ready for shipment to MSE-HKM Laboratory in a sealed cooler.

1.2.3 Sample Labeling

All samples will be clearly labeled following sample collection. The information recorded on the label will include:

- project name;
- demonstration site (ASARCO, Mineral Hill Mine, Montana Tech);
- sample type and analysis to be performed;
- sample description (i.e., influent, effluent):
- date and time sample was collected;
- sampler's initials;
- sample identification number;
- preservative and/or sample preparation techniques (i.e., preservatives added, filtered); and
- other remarks or special instructions.

To ensure that each sample is assigned a unique sample identification number, the following information will be provided about each sample:

- process identification [mineral-like precipitation (ML), ferrihydrite (IC), or alumina adsorption w/microfiltration (AL), stability tests (S)];
- demonstration site [ASARCO (A), Mineral Hill (M), Montana Tech]; and
- sample number (samples will be numbered consecutively from 101 for mineral-like precipitation, from 201 for ferrihydrite, from 301 for alumina adsorption w/microfiltration, and from 401 for stability tests).

For example, the first sample collected from the mineral-like precipitation process during the demonstration at the ASARCO East Helena lead smelter would be assigned the following sample identification number: MLA-101. Each sample following this collection would be assigned the next consecutive number (i.e., MLA-102, MLA-103, etc.)

2. ANALYTICAL PROCEDURES AND CALIBRATION

The analytical procedures used during the project are summarized in Table A-3. A brief discussion of the procedures used for each analysis are discussed below.

Table A-3. Project Schedule and Milestones for MWTP, Activity III, Project 9.

Milestone	Date
Work Plan	February 28, 1997
Site Access Agreements	May 30, 1997
NEPA Documentation	May 16, 1997
Vendor Subcontracts	June 30, 1997
Preliminary Design	March 21, 1997
Definitive Design	June 30, 1997
Project Specific Quality Assurance Project Plan	June 30, 1997
Fabrication	July 15, 1997
Construction Subcontract	June 20, 1997
Field Installation	July 18, 1997
Finalize Test Plan	July 4, 1997
Field Demonstration Completion	October 31, 1997
Oraft Final Project Report	January 30, 1998
Stability Testing	October 31, 1999
Revision to Final Report	Nevember 30, 1999

2.1 DISSOLVED ARSENIC (AA)

A Varian-Spectra AA 400 Graphite Furnace AA with Zeeman background correction will be used to analyze the As speciation vials as well as the dissolved arsenic concentration in the effluent at the MSE-HKM Laboratory. The furnace AA will be calibrated according to procedures outlined in EPA SW-846 Method 7060.

2.2 DISSOLVED, TOTAL RECOVERABLE, AND TOXICITY CHARACTERISTIC LEACHING PROCEDURE METALS ANALYSIS BY INDUCTIVELY COUPLED PLASMA EMISSION SPECTROMETER

Dissolved and total recoverable metals will be determined using SW-846 Method 6010A on an ARL 3560 Inductively Coupled Plasma Emission Spectrometer (ICP). The samples will be prepared for ICP analysis as outlined in SW-846 Method 3005A. The digestion procedure for total recoverable metals

will be modified to result in a matrix of 1% rather than 2% nitric acid.

The ICP will be calibrated according to the procedures outlined in SW-846 Method 6010 and the equipment manufacturer's instructions. Calibration will consist of the following procedures and items:

- mixed calibration standards;
- calibration blanks and reagent blanks;
- independent check standard;
- interference check solutions; and
- quality control samples.

2.2.1 Initial Calibration Verification

Calibration of the instrument will be verified using a mixed calibration standard from a different source and at different concentrations than the calibration standards. The concentrations of the analytes will be within the calibration range of the instrument and the analytes must be between 90%-110% of the true value or the calibration of the instrument will be repeated.

2.2.2 Continuing Calibration Verification

The calibration of the instrument will be continuously monitored by analyzing a CCV every 10 samples or every two hours, whichever is more frequent. The limit of acceptance for this standard is also 90%–110% of the true value of the analyte concentration. Should the calibration be out of compliance, the instrument will be recalibrated and the samples analyzed since the last compliant CCV will be reanalyzed.

2.2.3 Initial and Continuing Calibration Blanks

An ICB will be analyzed immediately after the ICV, and a CCB will be analyzed following the CCV. If the absolute value of the concentration exceeds the instrument detection limit, the instrument will be recalibrated and any samples since the last compliant CCB will be reanalyzed.

2.2.4 Interference Check Samples A and B

To verify interelement and background correction factors, an ICSA and ICSAB will be analyzed at the beginning and end of each analysis run or at a minimum of twice per 8-hour working shift. Results should fall within the control limits of 20% of the true value. If the results fall cutside the specified control limits, the analysis will be terminated, the problem corrected, the ICP recalibrated, and the analytical samples reanalyzed since the last compliant interference check sample will be reanalyzed.

2.3 pH

Although process pH measurements will be made through installed probes, some pH measurements will be done manually using a hand-held probe. A pH meter with automatic temperature compensation

capable of measuring pH at the demonstration site to 0.1 pH units will be used for this project. The pH probe will be calibrated daily using two fresh buffer solutions that bracket the expected pH. The meter will be calibrated before analysis begins, and calibration will be verified following the initial calibration and every 10 samples using a third buffer solution within the calibration range. If the third buffer solution differs from the true value of the buffer by more than 0.2 pH units, the meter will be recalibrated, and all samples analyzed since the last compliant calibration verification will be reanalyzed.

2.4 EH

An Orion En meter with a silver/silver chloride reference electrode will be used to determine the En at the demonstration site. The electrode will be calibrated using Zoebell's solution of known En. The measured Zoebell En must be within 20% of the known solution value or the probe will be subjected to cleaning and other required maintenance before recalibration.

2.5 ARSENIC SPECIATION

The total dissolved As, As⁺³, and As⁻³ in influent and effluent samples from the process trains will be determined using furnace AA, following the speciation using the Ion Exchange Ficklin Method (Ref. 1). The SOP that the MSE-HKM laboratory will be using is contained in Appendix C. The procedure involves passing 5 mL of the filtered, acidified sample through an ion exchange column packed with Donwex 1 x 8 anion exchange resin in 100-200 mesh size. The As+5 adheres to the acetare form of the ion exchange resin, while the As+3 passes through the column. To ensure the recovery of all of the As+1, the column is eluted with three separate 5-mL portion of deionized water. The original 5-mL sample and each elution will be collected in separate vials numbered 1 through 4. These vials contain the As+3. The column is then eluted with three separate 5-mL portions of 0.12M of Hydrochloric Acid (HCL). The pH change and the subsequent ion exchange cause the As+5 to pass through the column. The final three vials of sample collected contain the As+2. All of the speciation vials, as well as an unspeciated total dissolved arsenic sample, will be analyzed by furnace AA to determine the concentrations of total dissolved As, As⁻³, and As⁻⁵. The concentrations of the species added together divided by the measured total dissolved concentrations of arsenic will be calculated to determine if the recovery is acceptable (80%-120% recovery). If the recovery is not acceptable, the analyses will be repeated. This is a specific calculation for arsenic speciation analysis and should not be confused with the spike recovery calculation. The percent recovery of arsenic calculation will be determined using the calculation presented in Section 9 of the QAPP.

2.6 IRON SPECIATION

The concentration of dissolved iron will be determined by ICP at the MSE-HKM Laboratory. The concentration of ferrous iron will be determined using the colorimetric Standard Method 3500-Fe D and phenanthroline as the color developer. The spectrophotometer will be calibrated with a blank and at least three standards. The concentration of ferric iron will be calculated by subtracting the concentration of ferrous iron from the dissolved iron concentration determined by ICP. The MSE-HKM Laboratory SOP for this analysis is contained in Appendix C of the OAPP.

2.7 TOTAL SUSPENDED SOLIDS

Total suspended solids (TSS) will be determined at the MSE-HKM Laboratory according to EPA Method 160.2. A homogenized sample will be filtered through a glass fiber filter and the residue retained by the filter is dried to a constant weight in a 103-105 °C oven. A duplicate and a blank sample will be analyzed every 10 samples. Calibration of scales and oven temperature are verified on a daily basis and recorded in laboratory notebooks.

2.8 SULFATE

A Perstorp Flow Solution Auto Analyzer will be used to make sulfate determinations at the MSE-HKM Laboratory according to EPA Method 375.2. The auto analyzer is calibrated using at least five calibration standards between 10 and 200 mg/L. The calibration curve is then verified by analyzing an initial calibration verification standard, which is from a different source than the calibration standards. Calibration is continuously verified by analyzing a continuing calibration standard every 10 samples. Initial and continuing calibration blanks are also analyzed to verify that no significant contamination will occur at the instrument from instrument carryover from the calibration standards.

2.9 TOXICITY CHARACTERISTIC LEACHING PROCEDURE

Solid materials will be subjected to the TCLP procedure outlined in SW-846 Method 1311 at the MSE-HKM Laboratory. If sufficient sample is not available from filter cake samples, the TCLP procedure will be modified according to the weight of the solids submitted for analysis. The amount of extraction fluid added to the sample is determined by the weight of the sample and will be adjusted according to the sample weight. All reagent additions will be adjusted accordingly. The resulting extraction fluids from the TCLP will be digested according to procedures outlined in SW-846 Method 3005A for total recoverable metals. In addition to the reagents listed in the method 20 mL of 30% H₂O₂ will be added to the samples to help degrade the acetic acid. Digested samples will be analyzed by ICP. The ICP will be calibrated as discussed in Section 4.1 of the QAPP.

2.10 TOTAL METALS

The solid samples will be characterized for total metals by ICP SW-846 Method 6010A at the MSE-HKM Laboratory. Samples will be digested according to SW-846 Method 3050A. The ICP will be calibrated as discussed in Section 4.1 of the QAPP before sample digestates are analyzed.

2.11 PERCENT SOLIDS

The percent solids of each solid sample will be determined at the MSE-HKM Laboratory using the method outlined in Exhibit D, Part F of the Contract Laboratory Program Statement of Work, Document Number Ilm03.0. A copy of this method is contained in Appendix C of the QAPP. The percent solids data will be used to report the to metals on a dry weight basis.

2.12 STABILITY TESTS

In order to determine if the filter cakes generated by mineral-like precipitation and ferrihydrite are stable to tailings pond environment using the Montana Tech Procedure, a long term leach test developed during MWTP Activity 4, Project 5. The stability testing will be performed at the Metallurgy Department at Montana Tech. The procedure for each stability test is presented in Appendix B. The tests will be performed in triplicate on the filtercake samples generated during the project.

2.13 X-RAY DIFFRACTION

X-ray diffraction analysis will be performed using a Phillips 3100 X-ray Generator. The instrument will be calibrated suing a National Institute of Standards and Technology standard reference material. If the average delta 2-theta is greater than 0.1, the instrument will be recalibrated. Sample preparation procedures are discussed in the SOP contained in Appendix B of the QAPP.

2.14 CALIBRATION PROCEDURES FOR FIELD PROCESS INSTRUMENTS

All instruments used to measure field process variables will be calibrated using National Institute of Standards and Testing (NIST) traceable test equipment furnished by the Western Environmental Technology Office (WETO) in the MSE Technology Applications Instrumentation and Control Laboratory. The test equipment used during the calibration procedures are verified on a routine basis at Instrument Repair Laboratory in Broomfield, Colorado. This is a certified NIST secondary standards laboratory. Calibration procedures for critical and noncritical process field measurements are summarized in Table 18 of the QAPP. Calibration procedures will be performed prior to project startup and after the project has been completed.

Table A-4. Sample port/location descriptions and sample matrix at each location for the minerallike precipitation skid and the ASARCO ferrihydrite process.

Sample Port/Sample	Description	Matrix		
101	Process influent for the mineral-like precipitation system	aquecus		
102	Process water after H3FO4 addition for the mineral-like precipitation system	aquecus		
104	Process water after seed and CaC addition for the mineral-like precipitation system	aquecus		
106	Treated water discharge for the mineral-like precipitation system			
Filter Cake #1	Fl Sludge product from the mineral-like precipitation system			
FIT	Flow totalizer in the mineral-like precipitation system			
ρΗ	pH Tank 101, 102, 104 pH moritors in the mineral-like precipitation system			
401	40. Process influent for the ASARCO ferrihydrite process			
406	Treated water discharge for the ASARCO ferrihydrite process	aqueous		
Filter Cake #4	Sludge product for the ASARCO ferrinydrite process	solid		

Table A-5. Noncritical and critical measurements and frequency for the demonstration of the mineral-like precipitation process and the ferrihydrite process at the East Helena ASARCO lead smelting plant

Measurement	Matrix	Classificatio n	Sample Frequency	Sample Location	Total Number of Samples
рН	aqueous	roncritical	Initially, every hour for 8 hours, then every 4 hours	pII probes in tank 101, tank 102, and tank 104	69
рН	aqueous	ronari.ical	Before discharge	106	3
Ен	aqueous	roneritical	Initially, every hour for 8 hours, then every 4 hours	102 and 104	46
Total flow	aqueous	roneritical	Initially, every hour for 8 hours, then every 4 hours	FIT (total flow indicator)	23
As spec ation	aqueous	noncritical	Every 24 hours of operation	101 and 106	6
Iron speciation	aqueous	noncritical	Every 24 hours of operation	101 and 106	6
Tetal recoverable metals (As, Cd, Cu, Pb, Fe, P, Zn)	aqueous	noncritical	Every 24 hours of operation	101, 102, 104, 106	12
Disselved metals(Cd, Cu, Fe, Pb, 7n)	aqueous	noncritical	Every 24 nours of operation	101, .02, 104, 106	12
Dissolved metals(As, P)	aqueous	noncritical	Initially, every 12 hours of operation	102 and 104	14
Total metals (As, Ba, Cd, Ct, Cu, Fe, P, Pb, Sc, Ag, Zn, Ca)	solid	noncritical	Each sludge sample	Filter Cake #1	3
% solids	solid	noncritical	Each studge sample	Filter Cake #1	3
TCL? (Ba, Cd, Cr, Pb. Hg, Sc, Ag)	solid	noncritical	Each sludge sample	Filter Cake #1	3
Stability tests	solid	noncritical	Each sludge sample	Filter cake #1	3
TCLP (As)	solid	critical	Each sludge sample	Filter cake #1. ASARCO sludge	3
D ssolved metals (As)	aqueous	critical	Initially, every 8 hours of operation	101 and 106	18

Table A-6. Noncritical and critical measurements for the ASARCO ferrihydrite process.

Measurement	Matrix	Classification	Frequency	Location	Total Number of Samples
μH	aqueous	noncritical	initial	40: and 406	2
En	aqueous	noncritical	initial	401	1
Flow rate	aqueous	noncritical	initial	401	1
As speciation	aqueous	noncriteal	initial	406	1
Total recoverable metals (As, Cd, Cu, Pb, Fe, Zn)	aqueous	noncritica	initial	401	1
Dissolved metals (Cd, Cu, Fe, Pb, Zn)	aqueous	зунктіціа	in:tia.	401	1
TCLP (Ba, Cd, Cr, Pb, Hg, Se, Ag)	sclid	noncritical	once	Filter cake #4	1
TCLP (As)	solid	critical	once	Filter cake #4	1
Dissolved metals (As)	aqueous	crtical	Every Jay the mineral-like precipitation process is being demonstrated	401 and 406	6

Note: Initial samples will be collected from the ASARCO ferribydrite process just prior to the startup of the mineral-like precipitation process demonstration.

Table A-7. Noncritical and critical measurements for the mineral-like precipitation process demonstration at the Mineral Hill Mine (17-day test with 1,300-foot-level portal discharge water).

Measurement	Matrix	Classification	Sample Frequency	Sample Location	Total Number of Samples
pН	aqueous	noncritical	Initially every hour for 8 hours, then every 4 hours	pH probes in tank 101, tank 102, and tank 104	144
ρН	aqueous	noncritical	Before discharge	106	11
Вп	aqueous	noncritical	Initially every hour for 3 hours, then every 4 hours	102 and 104	96
Total flow	aqueous	noncritical	Initially every heur for 8 hours, then every 4 hours	FIT (:otal flow indicator)	48
As speciation	aqueous	noncritical	initial	1C1	1
Total recoverable metals (As, Cc, Cu, Pb, Fe, P, Zn)	aqueous	noncritical	Every 48 hours of operation	101, 102, 104, 106	16
Dissolved metals (Cd, Cu. Fe, Pb, Zn)	aqueous	noncritical	Initially every 48 hours of operation	101, 102, 164, 106	16
Dissolved metals (As, P)	aqueous	noncritical	Every 24 hours of operation	102 and 104	16
Total metals (As, Ba, Cd, Cr, Lu, Fe, F, Ph, Se, Ag, Zn, Cs)	solid	noncritical	Each sludge sample	Filter cake ≠1	1
% Solids	solid	noncritical	Each sludge sample	Filter cake #1	1
TCLP (Ba, Cc, Cr, Pb, Hg, Se, Ag)	solid	noncritical	Each sludge sample	Filter cake #1	1
Stability tes:s	solid	noncritical	Each sludge sample	Filter cake #1	1
TCLP (As)	solid	eritical	Each shudge sample	Fire: cake #1	1
Dissolved metals (As)	aqueous	critical	Initially every 8 hours of operation	101 and 106	44

Table A-8. Noncritical and critical measurements for the mineral-like precipitation process with concentrated arsenic brine from the alumina adsorption with microfiltration process.

Measurement	Matrix	Classification	Frequency	Location	Total Number of Samples
рН	aqueous	noncritical	Initial, every 4 hours	pII probes in tark 101, tark 102, and tark 103	12
pH	aqueous	noncritical	Before discharge	106	1
Ен	aqueous	noncritical	Initial, every 4 hours	102 and 104	8
As Speciation	aqueous	noncritical	Initial	101	1
Total Recoverable Metals (Al, As, Cd, Cu. Pb, Fe, P. Zn)	aqueous	noncritical	Initial	101 and 105	2
Disselved Metals (Al, Cd. Cu. Pb, Fc, P. Zn)	aqueous	noncritical	Initial	101 and 106	2
Disselved Metals (As)	aqueous	critical	Initial, every 4 hours	101 and 106	6

Table A-9. Sample port/location descriptions and sample matrix at each location for the ZENON

alumina adsorption with microfiltration process.

Sample Port/Sample Location	Description	Matrix
301	Process water influent	aqueous
302	Adsorption reactor effluent	aqueous
304	Treated water discharge	aqueous
305	Adsorption reactor tank	aqueous
306	Settling tank	aqueous
Filter cake #3	Sludge product	solid
FIT	Flow totalizer	aqueous

Table A-10. Noncritical and critical measurements for the alumina adsorption with microfiltration demonstration at the Mineral Hill Mine (1 test with 1,300-foot-level portal water, and tests with effluents from other processes, if necessary).

Measurement	Matrix	Classification	Sample Frequency	Sample Location	Total Number of Samples
pH	aqueous	noncritical	Initially, every 12 hours	301, 302, 303, and 304	52
pH	aqueous	noncritical	Before discharge	304	ć
Total flow	aqueous	noncritical	Initially, every 12 hours	FIT (Total flow indicator)	13
Total flow	aqueous	norcritical	initial, every 4 hours	FIT	4
Total suspended solids (TSS)	aqueous	norcritical	Initially, every 24 hours of operation	301, 302, 305, and 305	28
Sulfate	aqueous	norcritical	Initially, every 24 hours of operation	301 and 302	8
Total recoverable metals (Al. As, Cd, Cu, Pb, Fe, Zn)	aqueous	noncritical	Initially, every 24 hours of operation	3C1, 302, and 304	8
Dissolved metals (Al, Cd, Cu, Fe, Pb, Zn)	aqueous	noncritical	Initially, every 24 hours of operation	301, 302, and 304	8
Dissolved metals (As)	squeous	critical	Initially, every 12 hours of operation	301 and 304	26

Note. Initial samples will be collected after one system volume has been processed. Total number of samples does not include samples from processing effluents from the other processes since it is unknown at this time if further processing of the effluents will be necessary.

Table A-11. Sample port/location descriptions and sample matrix at each location for the ferrihydrite process at the Mineral Hill Mine.

Sample Port/Sample Location	Description	Matrix	
201	Process influent	aqueous	
202	Process influent after FeCh addition	aqueous	
204	Process influent with HCl and CaO addition	aqueous	
206	Treated water discharge	aqueous	
Filter cake #2	Sludge product	solid	
FIT	Flow totalize:	aqueous	
pН	Tank 201, 203, and 201 pH monitors	aqueous	

Table A-12. Noncritical and critical measurements for the Mineral Hill demonstration of the ferrihydrite process (day test with 1,300-foot level portal discharge water).

Measurement	Matrix	Classification	Sample Frequency	Sample Location	Total Number of Samples
pH	aqueous	noncritical	Initially, every hour for 8 hours, then every 4 hours	pH probes in tank 201, tank 203, and tank 204	126
pН	aqueous	noncritical	Before discharge	206	9
Вн	aqueous	noncritical	Initially, every hour for 8 hours, then every 4 hours	202 and 204	84
Total flow	aqueous	noneritical	Initially, every hour for 8 hours, then every 4 hours	FIT (total flow indicator)	42
Total recoverable metals (As, Cd, Cu, Pb, Fc, Zn)	aqueous	noncritical	Every 48 hours of operation	201, 202, 204, 206	12
Dissolved metals (Cd, Cu, Pb, Zn)	aqueous	noneritieal	Every 48 hours of operation	201, 202, 204, 206	12
Dissolved metals (As, Fe)	aqueous	noncritical	Initially, every 24 hours of operation	202 and 204	14
Total metals (As, Ba, Cd, Cr, Cu, Fe, Pb, Se, Ag, Zn, Ca)	bilcs	noncritical	Each sludge sample	Filter cake #2	1
% solids	solid	noncritical	Each sludge sample	Filter cake #2	1
TCLP (Ba, Cd, Cr, Pb, Hg, Sc, Ag)	biles	noncritical	Each sludge sample	Filter cake #2	1
Stability tests	biles	noneritical	Each sludge sample	Filter cake #2	1
TCLP (As)	solid	critical	Each sludge sample	Filter cake #2	1
Dissolved metals (As)	aqueous	critical	Initially, every 8 hours of operation	201 and 206	38

Table A-13. Noncritical and critical measurements for the ferrihydrite process using concentrated

arsenic brine from the alamina adsorption with microfiltration process.

Measurement	Matrix	Classification	Frequency	Location	Total Number of Samples
pН	aqueous	neneritical	Initial, every 4 hours	pH probes in tank 201, tank 203, and tank 204	12
pH	aqueous	noncritical	Before discharge	206	1
En	aqueous	noncritical	Initial, every 4 hours	202 and 204	3
Total recoverable metals (Al, As, Cd, Cu. Pb, Fe, P, Zu)	aqueous	noncritical	Initial	206	1
Dissolved metals (Al, Cd, Cu, Pb, Fe, P, Zn)	aqueous	noncritical	initial	206	1
Dissolved metals (As)	aqueous	critical	initial, every 4 hours	201 and 206	6

Table A-14. Field quality control sampling for each process demonstration.

Analysis	Field Duplicates	Field Blanks
Dissolved Arsenic	-11	I ¹
TCLP	12	N/A

Field QC samples are to be taken at the initial and final sampling events technology demonstration. The initial set of field QC samples will be taken from influent locations (101, 201, and 301), and the final set of field QC samples will be taken from the effluent locations (106, 206, and 304).² A Field duplicate will be taken at each site for each technology demonstration from resulting filter cakes.

Table A-15. Sampling frequency for the stability tests.

Measurement	Matrix	Classification	Frequency	Sample Location	Total Number of Samples
Sample weight	solid	noncritical	Before and after test	Sample before test, vacuum filter after test	48
Dissolved As	aqueous	noncritical	Every year	Syringe filtrate	24
pH	starry	noncritical	Every year	Reaction vessel	24
En	slarry	noncritical	Every year	Reaction vessel	24
Temperature	slarry	noncritical	Every year	Reaction vessel	24
X-ray diffraction	solid	noncritical	After test	Vacuum filter	12

Table A-16. Total number of samples.

Analysis	Primary Samples Site Demos	Primary Samples Stability Tests	Field Duplicates	Field Blanks	Project Total
ρH	447	24	n'a	n/a	471
En	2/13	24	n/a	n/a	267
Total flow	134	n/a	n/a	n/a	134
TSS	28	n/a	n/a	n/a	28
Sulfate	8	n/a	n/a	n/a	8
Arsenic speciation	9	n/a	n/a	n/a	9
Iron speciation	6	n/a	n/a	n/a	6
X-ray diffraction	nva	12	n/a	n/a	12
Total recoverable metals (aquecus)	52	n/a	n/a	n/a	52
Dissolved metals	101	n/a	n/a	3/8	101
Total metals (solid)	5	n/a	n/a	3/8	5
TCLP	6	n/a	6	1/3	12
Dissolved arsenic	144	24	12	12	192
% solids	5	n/a	n/a	n/a	5
Total number of samples	1,188	84	18	12	1 302

2.14 REFERENCES

 Ficklin, Walter H., "Separation of Arsenic(III) and Arsenic(V) in Ground Water by Ion Exchange," Talanta Vol 30, No. 5, 1983, pp 371-373.

APPENDIX B

Test Data

	(III) As At (I) At (As II)																			2982			
Arean Speciation Results	As (10)																			787			
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	+ (<u>p</u>																E E	59,256		10.753			
	is (10)																7000011						
4	3 %																27,333	27.93	-	3,316			
MINERAL-LIKE PRECIPITATION TEST DATA	F (mg/l)		200'00'	60,758	75 967	651 150	34.235	78 M		7709.83	14.343	11.462					6.00	90.89	8.981	9		77,295,056	121.47
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T T	Cu (regit)																		900	02 40,250			
ALL	Chart.																104.47	195.313	E 176,539	38:.102			
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≥ ٍ	(mol.)			C.05272			C.02595	C30307	0.671			500.00		0.09532		0.00774				0.00613	D.CC772		0.2723
7.	(CP)		2093 783	2000	3152	3151 072	2000	0.0399		34773	0.637	0.053	0316414	00473			3276.836	21.81.9	0.144	0.00878		3343,132	0,0052
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	CELO	M_A-2104	M-3,511	MJA 212	M.A.213	M.A.214	M.A-215	M_A.216	M.A.2169	M.A.2163	MLA 2160	MASIG	Mineral Re Process Demonstration (Mineral Hill Mater)	M_M-300	10.70	M_M-302	M. M 303	M_W-304	M W-30F	M.W-00C	M.VA307	M W III	DI-98-962	M_W-305A,	M WATE	M_W3*1	M.M.372	MARY	M. U.S. 6	24 14 21 2	

			24 (C)		\$ 55 \$														284.02										
			Age (III) (Ug/LI		8														88										
			A serio Specialus Plesa Total Urato As As 1991, (USA) (U		354,3														396.3										
			A Section Sulfate Impli-																										
			Chards (1991),																										
	0.002	6,613	75 (J.GE)	110.2	40.05R										390'3	C.04.5	5.00	-0.000											
	0.048		т (лап)	0.040											0,443		0.028												
			3 (100.0)																										
	0.038		Se (mgk.)	0.035											0.048		0.023												
4003	69000	50 C3	9	0.002	4310		0.002		0.047	000		6(2)0	4000	0.003	6969	0.381	0.403	40,310			4019	0.021	0031		A0.00	0.058	2000	0035	2000
	0.321		MA (mgr.)	6.307											8100		1000												
	<0.02	<0.02	% (Mg/L)	-0.02	-C206										-0.02	9,02	20105	-0.256											
	C.57E <0.02	40.034	7.6 (TQU)	25 E	00342										350'3	+0.024	C.028	40.242											
	0.008	9000	CJ.GUI)		800										-0.003	0.003		-0.020											
36 360	123.354	124.930	S Co		301.238										118.302 -0.003	119,813	202	214.020											
	1000	3,005	nau Ca		6(CU)											4000	3009	9000											
0.00019			As (M) (D)U		0,00629	DOD'N.	0.01063		60,0000	0.00073	100000	000007	a manual	0.00000			TO COME	0.00441		0.00713		0.00	0.00419		0.00000	2000	0,00102	1000	000157
40315	0.445	0.423	/e (ICF) IMp()		2 GH (2		6.324	0.458	5000	0.488	0.305	0.305	6,50	0.020	0.42	0.447		+10.0			17'0	6000			0.00				0.00
	0.023		N (TCL)	27											2000		20100												
OISS	TREC	DISS	Sample 17PE	TREC	DISS	0188	CNSS	DISS	SSIC	083	SSKO	USS	MSS	OSO	TREC	DISS	CMSS	000	SPEC	OISO	OSS	888	OISS	SSIC	388	JESS	286	SSIC	255
1000AM	OSCO PM	03 CO PM	Time	MAGOSO	OBCO PM OBCO PM	MACOST	OSCOPIA COCOPIA	05 CD PW	Md op so	02 CO AM	02 CO AM	OZCO AM	11CO AM	1300 AM	13C0 AM	02 CO P.M	02:00 PM	02.00 P.M	02:00 PM	NJ 03/20	02:00 PM	Men or or	02:00 PM	MG DOOD	TOTO PM	TEXTOAM	DOLCO AM	DECO AM	OR SO AM
08/16/97 1	0.097697 0	00/15/97 0	Dale Sampled 3		08/16/97 0 06/16/97 0		09/16/95			0.09/17/97 0				00/17/87	1 (0/11/60)	09/17/97 0	0 /6/11/60		0.08/17/97 0	00/17/95 0		CHANGE OF		787787					Certery, C
1050	10.	10. 0	Eample Port 2		0 0	-	8 8			2 5				1020	10:	100	1080		9 6			1 0	0	101					105
MUM-225 MUM 826	MLM-327	BCC-W1W	FIE.D 4	MUM-229	MLM-330 MLM-221	MIMES	MUMOSSA	MLM-235	MLM-336	MLM-227	MLM 336	MLM-340	MIMSH	MLM-042	MLM 343	MLM:344	MLM-345	MLM:040	MLM:347 MLM:349	MLMCAB	MLM350	Miller FT	MLM363	MILITERS	MLMSS	MI MASS	MLM-3C0	MLM-261	MLANCE

and the same of																					
HELD	Sample	Date	E C	Sample A	₹ = .			6			2	Pa Min	Mn o	22	E	52	Sufface	EST]	As As) (S.)	As CA
ID Alamina Abs	Pur.	Sampler, w Level Are	Serroted and Demon	year	 Pur Samplet, Sembled Tyter (myt.). Alamina Absorption Low Level Aryanic Demonstration (ASARCO) 	ngC) (mg/c	(195)		then) th						1.61	Taffur)	(1181)	(MB/C)	0.000	0.64	(49/1.
PLA1-101A	3014	067-4/97	067-4/97 DELIS AM	SFEC															9.44	881.75 <20	62
PLA1-103	301	266.393	X 10 PM	TEBC	000	2.230	0	0.204	727.09	9,854	323	251.3	9,142		7.441						
PLA1-103A	301	067.387	NETO PM	DISS	Ø 00	2.1%	0	-		0.051	3.153	5000	7,351		7 196						
PLA1-103B	3018	20/2-790	MH OTH	TREC	40.00	2.252	0		747 392	6,618	3217	187	271. 6		7.598						
PLAT-103G	3018	180, 297	24.10 P.M	DBS	20.9	2.155	0.			120.0	3.122	C.104	1,271		0.901						
PLA1-103D	3014	1667.790	04:20 PM	THEC	900	1.86	0.	0.184 72	727.298	0.012	3.175	L.113	8.345		7.546						
PLA1-103E	3014	TM2-190	M-20 PM	DISS	DIE DIE	0.524	Ų	89 8-10	698 152	0.64	2002 4000	0	A5E 0		6673	0.015					
PLA1-104	306	1667.397	X 35 PM	T88														25/8	50		
PLA1-105	302	667.397	34.30 PM	DISS	0,029	1303															
PLA1-105	302	2667.307	MASS PM	DES	C.481	1.356															
PLAI-107	302	1872 /30	MASSEM																		
PLA1-103	305	060-393	35:12 PM	DISS	69.3	0.972															
PLA1-109	302	(6/5-/30)	35d2 PM																		
PLAI-113	303	DEC 2597	35.30 PM	DBS	20.05	1,578															
PLA1-111	301	067:397	35.30 PM	DBS	2000	6F".															
PLA1-112	304	20/5 //39	05:30 PM																		
PLAI-113	302	287. 237	35:30 PM	DESS	C.032	1,022															
Pt 41.114	3002	587:790	NG 00-30																		
PC.84.115	30.1	(0/5-/30)	N-10 PM	DISS	- QUUP	1.500															
PLAIL IIS	301	067.2/87	N 26.50	-																	
PI A1.117	306	097:5/07	NG 00-30	138														2333	3		
PLA1-113	302	067-5/87	26:30 PM	DES	7,708	1,518															
PLAI-113	302	067,3397	36:30 PM																		
Pt 41.123	306	587:380	38-30 PM	Tas														2453	100		
PLA1-121	302	087:3/80	38:30 PM	DISS	12,219	1,31															
PLAI-122	302	105/3/3/	25.30 PM																		
PLA1-123	305	08/13/97	10:30 PM	138														2378	m		
PLA1 124	300	08/12/92	16:30 PM	SSIG	30,63	3.14															
PLA1-125	305	D8/12/97	10:30 PM																		
PLA1-125	301	087:4/97	12:30 AM	DISS	40.02	1,506															
PLA1-127	30.1	08/1/92	12:30 AM																		
PLA1-125	300	76/2/1/00	12:50 AM	155														1261	- Are		
PLA1-129	305	160/14/97	12:30 MM	SSIG	2.37	1335															
PL/11 133	3005	79/1/90	12:30 AM																		
PLP1-151	5005	VEDATINO.	12:30 AM																		
PLAT-152	306	08/14/97	32:30 AM	T38														2340	0		
PL/11 133	302	00/11/07	32:30 AM	SSId	12.674	1256															
PLP1-139	305	7874780	D2:50 AM																		
PLA1-135	305	08/14/57	SHISS AM	138														2271	-		
PLA1 135	306	08/11/97	D8:30 NM	139														202	00		
PLAT-TS/	305	76771/90	J8:30 AM	1991														2745	n		
PLA1-138	300	78/7/80	38-30 AM	DISS	10.967	1.15															
DI A4 130	2002	00/1/00	30-30 ALA	990	45,435	4 643															
0000000	100	Ond street	20-20-514	8	4	1407															
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FLA1-141	2 3	16/41/00	10.30 AM	2 1														040	5 0		
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FB.0	Somole	Date	Time	Somolo A	A COURT	(HCF)	(30)	75	Co	30						-	5	Gulfale	32		As A
furrise Ats	Portion Pr	Oceass Low	campiec.	le Servor		SARCCI	(mbc)	Juger)	T BILL	(III)	(ngu)	(165)	(mgr)	(mg/c)	(morr)	mari	(4)%-1	Con	HIGH.	1	(nbr) (nbr)
RA1-101	391	(8.14,97	CB:14,97 35.3C PM	DSS	<0.204	2.279															
RJA1-100	305	CB114.97	35:30 PM	SSIC	6.43	C.213															
R.A.1-105	305	76,146,7	36:00 PM	SSIC		0.24	00000														
KAT-W	6.6	LK19/4	JESU PN	8 25		0.0275	-	(100)													
R.A.1.08	300	CB:14/97	NG 08:30	SSIC		0.0702															
KAN-UB	332	UB14/8"	NA DEVIC	2188		X200															
R. A11.10	30.	CB:14/97	DESTRUMENT	2188		0.035	6.05458														
RAFTI	3%	UBTRB!	MASKIC	8		0.221	ey.														
R.A5-12	370	IR1144*	Ne de a la	SSIC		273															
RAN IS	8 8	CB.14.97	2631 PN	886		0.0503	0.0000														
R. A. L. 15	37.5	C8.11/0"	Ne do sk	8 8		0.0000	2000														
RLASS 15	30	FR.11497	MI CESS	158	50.3E4	0.0243															
RLA1-18	302	08/14/97	36:32 PM	SSIC	00.204	0.0461	0. 1700														
Jurina Abe	orniton P	December Low	evo Arrest	ic Domen	Numina Abearation Process Low Level Arrents Domonstration (ASA/RCO)	SA BOOM															
10773H	.767	08/13/97	WY ODER	SSIC	40,204	0,0507	0.05512														
201 CYTE	30.	08/13/97	12:30 AM	Suifate														2127.5			
RLA2-103	305	308130	9230 AM	200	0.386	0.0000	0315														
71.42.104	302	08/15/07	12:30 AM	Suffate														2150,98			
ALAZ-103	202	00/13/97	MA 00.10	0000		1670															
N 42 100	100	Children	O PARTON	S S S		0.108	Low														
2142-103	305	084397	01-30 ZW	980		0.163															
3LA2-103	202	00043/97	02. B.AM	OBS		0.243															
31,42,113	30.	08/13/97	02:30 AM	OBS		0.0074	0.05372														
3LA2-111	200	2001202	WY 00:20	083		6,0019	0,3059														
3LA2-112	305	08/15/97	02:00 AM	OBS		0.294															
31/2113	304	08/13/07	W/ 00:50	0.65		0.101															
30/2/114	305	08016097	02:30 AN	0.65		0.276															
100.112	200	Carried Street	O-COLOR OF	200		OZD															
*LAZ-115	95	(Brist)	0430 AN	DES	40.004	0.0043	0.06004														
21/2/42	200	CRHSST	0450 38	Color	1000													214.31			
21/2/113	300	08/16/57	0450AN	Suffice	NOTA .	mon												2000,09			
britis Alas	ory Jun Pr	Ma" essar	Jasay Jasar	ic Demon	Akridi a Aksunyau Praces Lew Level Arseric Denombration (ASARCO)	PARCO															
MA-CI	301	CEMENT	DE CO AN	TREC	60.05	1023		0.238	0.238 1357.358	0.0	0.553	0.266	64,660			.2802	0.272				
ALP-101A	301	16/11/90	W/ 00:00	0155	50.02	0.36		131.0	0.151 1136.313	0.016	0.356 <0,02	<0.02	4.006			7.084	0.035				
ALA-103	301	08/16/97	06:00 AN	188															+	25	173
NO04	200	78/01/20	00.00 AN																		
ALA-105	388	08/16/N7	No.02:30	1.85															0099	2	0
70. ALA	581	CB/16/07	MK 00:30	Outste														2130,88			
ND-108	305	06/16/97	DE:00 AIV	Suffee														1845.49			
ALA-100	301	58/16/97 5	MV 0 280	DIBS		CF/0															
MAN-110	2007	US/16/9)	Me Disea	DISS		2000															
41.4.44						Special Street															

	3 0"31:	Samole	Sats	Time	Sample	W	AS (10 P.	(AA)				6		Mn P	a	F			Euffolo	183	138	As \$13	Ac (X
10 10 10 10 10 10 10 10	1	Por	Sempled	Sampled	lype .	(mg/l)	(mg/L)	(mort)											1	(mg/L)	(163	ngh.	(10g/L)
	150	200	08/897	D2 30 PM	2 2	700																	
	9 0	3 8	78/04/20	05.00 PM	252	-														-dr			
State Stat	114		72591.60	NG 05:30	TSS																		
State Stat	-115		06/16/97	03:30 PM	153																		
11 12 12 12 12 12 12 12	-117	100	20,91,80	MR 0030	5SIG		5533																
201 697709 110894 11085 10681 10889 10889 10889 10889 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 11089 1108	-118	305	18,91,50	US DU PM	CISS		0.238	1															
337 Graphy 113m M Title	110	301	08/18/97	MG 00711	CISS		0.0558																
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13 14 15 15 15 15 15 15 15	121	100	9017.07	DS:00 AM	TREC	-0.05	7.223	325	(257)	1007,050	3.32			E78.77			17,425	0.203					
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10 10 10 10 10 10 10 10	122A	305	76,71,80	MA 06:30	DISS	77.219			0.244				CD D	3F 037			10 397	0.108					
13.50	123	301	06317.90	MA 0050	155															228			
10 10 10 10 10 10 10 10	124	302	08/17/97	MA 0050	199																		
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	Sample Unio	-	***	Sample At Type (mpl.)	Ar (IICP) (mpd)	Ass. (Ass.)	(ms/.)	75 July	Ct.	Fe (mo/L)	Pt mol	Min (mp/L)	000	35	T (0.4.4)	Zr Oruto	Surfete (my/)	766 (1/4/0)	food As (up/L)	As 310 Guydo	S S S
AUM-198A			SOLD BY			936.0	40.004		0.0	DESTE AD DOG	STEE	-	100	Z		E					
03	707 DAD TOT	MARCON 180			0 54065		D 022575 N3 304	124 452	2 0009	600	1600	6.001 0.058 0.063	N 0,00	3 6,488	30	91.076					
91M-13UA	202 0872.097	DAY CASOPIN	PW DISS			0.075 0.02	D0223/ <0.004		33	2001 40,054	20.00		20,07			0.021					
	301 09/2/39/7	DAY CALCOPY		84														444			
	302 00/27/97	797 C4 00 PM	38_ MJ															- 35			
	that court has			21														17365			
	201 04/27/97	VALUE OF COMPA	PM Subset	1													787				
41N-135	365 08(2.19)7	W WORK	MA SUIDS	9													131				
Absero	Absorblion Precess Low Lavel Areane Correlation (ASARCO)	ow Level A	reanc Lam	onstration	MENROO																
31.44-101	301 00/27/97	292 CG 00 PM	P.M. DICS	d s	0.270 0	0.356															
	201 0027097	WAR COMPA		5													234				
TO 44-103	702 CMD 1/807	UST CREDITY	PW DISS	8 551.06		0.168															
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- SLAN-NUS	2002 CH22-1977			38	0	0.084 0.0	0.0051														
	301 0002797	727 CZ 00 P.VI	P.VI DICS	**	0	0.394															
	201 080 192			-	0	0.475															
NI 64-105	TRUTTOND CAR	DRY CREDING	PM DINS		•	0.387															
	202 08/2//97	NACE OF SUPPLY	PW DISS	34	n	0.335															
-	76777/RD 106			25	5	388															
	302 06/27/97	M7 0800 FW	P.M. DIES	-	0	212															
31.44.112	302 08/27/97				0	0.128															
	501 0807357	VEC COCD PV	PW DISS	-	0	378															
41.44-111	7607/80 TOS	MATORIA 160	PW DISS	20		0.0 1.0	0.0635														
40,44,115	302 08/2//07	me costin		26	n	0.00 J.00035	988														
	301 08/27/07		PM DIGS		0.17 0.	0.371															
	301 00/27/97		***	12													229				
S124.113	982 08/27/47	247 10.00 Ps	Pot DISS	8 177,198		0.054 0.05960	280														
SL44.119	902 08(27/9)	797 10:00 PV	PM Sanan	9													3114				

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					As	As							Total		
FIELD	Sample	Date	Time	Sample		(AA)	8	3	E.	Pp	۵	Zn	As	As (III)	As (v)
10	Port	Sampled	Sampled	Type	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(1/3n)	(ngd.)	(100)
Iron Copres	cipitation P	Iron Coprecipitation Process Demonstration (Mineral Hill Water	nonstration	(Mineral	Hill Water	-									
ILM-100-A	203A	09/02/97	12:15 AM	DISS	0.058	0.02742									
ILM-100-B	203A	09/02/97	12:30 AM	DISS	0.039	0.01704									
ILM-100-C	203A	09/02/97	12:45 AM	DISS	0.051	0.02018									
ILM-100-D	203A	09/02/97	01.00 AM	DISS	0.041	0.02153									
ILM-100-E	203A	09/02/97	01:15 AM	DISS	0.053	0.0228									
ILM-100-F	201	09/02/97	03:05 PM	DISS	0.419				0.155						
ILM-100-G	202	09/02/97	03:05 PM	DISS	0.02	0.00892			0.127						
ILM-100-H	202	09/02/97	03:20 PM	TM	0.37				6.316						
ILM-101-A	202	09/02/97	U6:10 PM	DISS		0.00596									
II M-101-B	202	78/20/80	DB-10 PM	SPEC									323.1		336.05
ILM-101-C	202	09/02/97	06:10 PM												
ILM-101-D	202	09/02/97	06:10 PM												
ILM-101	202	09/02/97	06:20 PM	SSIC	-0.005	0.00734			0.047						
ILM-102	204	09/02/97	06.20 PM	DISS	0.024	0.02837			<0.024						
ILM-103	204	09/02/97	06:20 PM	DISS	-0.038	<0.001									
ILM-104	201	09/02/97	06:20 PM	DISS	0.398										
ILM-105	203	09/02/97	06:20 PM	DISS	0.026	0.02261									
ILW-106A	201	09/03/97	01:20 AM	DISS	0.392										
ILM 106B	202	09/03/97	01:20 AM	DISS	0.01	0.0058									
ILM-106C	203	26/00/60	01:20 AM	DISS	0.073										
ILM-108D	204	09/03/97	01.20 AM	DISS	0.04	0.0319									
ILM-108	201	09/03/97	09:20 AM	DISS	0.384										
ILM-109	206	78/80/80	09:20 AM	DISS	0.072	0.03949									
II M-110	201	79/03/97	05:30 PM	DISS	0.363										
ILM-111	201	26/60/60	05:30 PM	DISS	0.379										
ILM-112	206	20/03/02	05:30 PM	DISS	0.056	0.05483									
ILM-113	202	76/60/60	05.30 PM	DISS	0.032	0.0107			0.031						
ILM-114	204	09/03/97	05.30 PM	DISS	0.00	0.03657									
ILM-115A	201	09/04/97	01:20 AM	DISS	0.385										
ILM-115B	202	09/04/97	01-20 AM	DISS	0.026	0.01841			<0.024						
ILM-115C	202	09/04/97	01:20 AM	TM	0.397				4.567						
ILM-115D	203	09/04/97	01:20 AM	DISS	0.049	0.04811									
ILM-115E	204	09/04/97	01:20 AM	DISS	0.04	0.04634									
ILM-116	206	09/04/97	01:20 AM	DISS	0.071	0.05232									
ILM-117A	201	09/04/97	09.30 AM	DISS	0.366										
11 14 1170	203	70/10/00	09:30 AM	Dies	0.004										

10 (1) (1) (1) (1) (1) (1) (1) (1) (1) (1)						As	As							Total		
Port Sampled Sampled Juge Impt. Im	FIELD	Sample	Date	Time	Sample	(ICP)	(AA)	3	S	Fe	Pb	۵	Zn	Ass	As (III)	As (V)
A 201 GRANGE CHASA ADDRA CDRA CDRA <th< th=""><th>ID CI</th><th>Port</th><th>Sampled</th><th>Sampled</th><th>lype</th><th>(mg/L)</th><th>(mg/L)</th><th>(mg/L)</th><th>(Ud/L)</th><th>(ma/L)</th><th>(mg/L)</th><th>(mg/L)</th><th>(mg/L)</th><th>(nd/L)</th><th>(nd/L)</th><th>(ng/L)</th></th<>	ID CI	Port	Sampled	Sampled	lype	(mg/L)	(mg/L)	(mg/L)	(Ud/L)	(ma/L)	(mg/L)	(mg/L)	(mg/L)	(nd/L)	(nd/L)	(ng/L)
A 201 0905/97 05:30 PM TREC 0.377 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.004 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044 0.0044	ILM-118	206	09/04/97	09:30 AM	DISS	0.044	0.0459									
201 09040497 6530 PM DISS 0386 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 0.004 <t< td=""><td>ILM-119</td><td>201</td><td>76/50/60</td><td>05:30 PM</td><td>TREC</td><td>0.377</td><td></td><td>600 U</td><td>0.007</td><td>0.141</td><td><0.02</td><td>0.044</td><td>0.031</td><td></td><td></td><td></td></t<>	ILM-119	201	76/50/60	05:30 PM	TREC	0.377		600 U	0.007	0.141	<0.02	0.044	0.031			
206 00040567 05:30 PM TREC 0.114 Annual 40.003 0.024 <0.024 <0.002 <0.014 206 0.004/407 06:30 PM DISS 0.046 0.07311 <0.0044	ILM 119A	201	09/04/97	05:30 PM	DISS	0.386		0.008	0.004	<0.074	<0.02	0.045	0.032			
206 0904/97 05:30 PM D1SS 0.049 0.07311 <0.004 <0.003 <0.0024 0.0022 206 0904/97 05:30 PM D1SS 0.045 0.0616 206 0904/97 05:30 PM D1SS 0.045 0.0616 206 0904/97 05:30 PM D1SS 0.045 0.0016 0.0016 0.0024 0.0024 207 0904/97 05:30 PM TREC 0.022 0.0416/0 0.0024 0.0027 0.0024 0.0024 0.0024 0.0024 0.0027 0.0026 PM D1SS 0.045 0.07140 0.0022 0.0424 0.0027 0.0026 PM D1SS 0.046 0.04941 0.0022 0.0424 0.0022 0.0424 0.0022 0.0042 0.0022 0.0424 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.0022 0.00	ILM 120	206	09/05/97	05:30 PM	TREC	0.114			<0.003	0.549	<0.02	<0.03	<0.000			
201 09004/07 05:30 PM DISS 0.38 206 09004/97 05:30 PM DISS 0.067 0.06456 202 09004/97 05:30 PM DISS 0.067 0.06456 202 09004/97 05:30 PM DISS 0.002 0.01496 202 09004/97 05:30 PM TREC 0.022 0.01496 202 09004/97 05:30 PM TREC 0.032 0.0713 202 09004/97 12:00 AM DISS 0.045 0.0713 202 09004/97 12:00 AM DISS 0.096 0.10961 202 09004/97 12:00 AM DISS 0.096 0.10978 203 09004/97 12:00 AM DISS 0.072 0.09471 204 09004/97 12:00 AM DISS 0.074 0.10728 205 09005/97 03:30 PM DISS 0.074 0.10728 206 09005/97 03:30 PM DISS 0.007 0.09601 207 09005/97 03:30 PM DISS 0.097 0.09601 208 09005/97 03:30 PM DISS 0.097 0.09601 209 09005/97 03:30 PM DISS 0.097 0.09601 200 09005/97 03:30 PM DISS 0.097 0.09601 201 09005/97 03:30 PM DISS 0.097 0.09601 202 09005/97 03:30 PM DISS 0.097 0.09601 203 06005/97 03:30 PM DISS 0.097 0.09601 204 09005/97 07:00 AM DISS 0.097 0.09786 205 05005/97 12:00 PM DISS 0.095 0.09786 206 05005/97 12:00 PM DISS 0.095 0.09786 207 05005/97 12:00 PM DISS 0.095 0.09786 208 05005/97 12:00 PM DISS 0.095 0.09788 209 05005/97 12:00 PM DISS 0.095 0.09788 200 05005/97 12:00 PM DISS 0.095 0.09788 201 05005/97 12:00 PM DISS 0.095 0.09788 202 05005/97 12:00 PM DISS 0.095 0.09788 203 06005/97 12:00 PM DISS 0.095 0.09788 204 05005/97 12:00 PM DISS 0.095 0.09788 205 05005/97 12:00 PM DISS 0.095 0.09788	ILM-120A	208	09/04/97	05:30 PM	DISS	0,049	0.07311	<0.004	< 0.003	<0.024		0.062	<0.00>			
206 09004/97 05:30 PM DISS 0.067 0.06456 206 09004/97 05:30 PM DISS 0.002 <0.001 202 09004/97 05:30 PM DISS 0.002 <0.01046 202 09004/97 05:30 PM TREC 0.022 0.01046 202 09004/97 05:30 PM TREC 0.392 2.010446 202 09004/97 12:00 AM DISS 0.045 0.0713 202 09004/97 12:00 AM DISS 0.096 0.10691 202 09004/97 12:00 AM DISS 0.096 0.10691 202 09004/97 12:00 AM DISS 0.097 0.09411 203 09005/97 03:30 PM TREC 0.374 0.10728 204 09005/97 03:30 PM DISS 0.097 0.09601 204 09005/97 03:30 PM DISS 0.097 0.09601 205 09005/97 03:30 PM DISS 0.097 0.09601 206 09005/97 03:30 PM DISS 0.097 0.09601 207 09005/97 03:30 PM DISS 0.097 0.09601 208 06005/97 03:30 PM DISS 0.097 0.09601 209 06005/97 03:30 PM DISS 0.097 0.09601 200 06005/97 03:30 PM DISS 0.097 0.09601 201 06005/97 07:00 AM DISS 0.097 0.09700 202 06005/97 07:00 AM DISS 0.097 0.09700 203 06005/97 07:00 AM DISS 0.097 0.09700 204 06005/97 07:00 PM DISS 0.097 0.09700 205 06005/97 07:00 PM DISS 0.097 0.09700 204 06005/97 12:00 PM DISS 0.097 0.09700 205 06005/97 12:00 PM DISS 0.097 0.09701 204 06005/97 12:00 PM DISS 0.097 0.09700 205 06005/97 12:00 PM DISS 0.097 0.09700 204 06005/97 12:00 PM DISS 0.097 0.09700 205 06005/97 12:00 PM DISS 0.097 0.09700 206 06005/97 12:00 PM DISS 0.097 0.09700 204 06005/97 12:00 PM DISS 0.097 0.09700 205 06005/97 12:00 PM DISS 0.097 0.09700	ILM-121	201	09/04/97	05:30 PM	DISS	0.38										
206 0904/97 05:30 PM DIBS -0.022 -0.0014 202 0904/97 05:30 PM DIBS -0.022 -0.01603 202 0904/97 05:30 PM TREC -0.022 0.01846 204 0904/97 05:30 PM TREC -0.032 0.0164 202 0904/97 05:30 PM TREC -0.392 202 0904/97 12:00 AM DIBS -0.045 0.07113 202 0904/97 12:00 AM DIBS -0.096 0.10491 202 0904/97 12:00 AM DIBS -0.096 0.10491 202 0904/97 12:00 AM DIBS -0.097 0.09691 203 0904/97 12:00 AM DIBS -0.097 0.09691 204 0905/97 03:30 PM TREC -0.374 0.10728 204 0905/97 03:30 PM TREC -0.374 0.10728 204 0905/97 03:30 PM DIBS -0.017 -0.09447 204 0905/97 03:30 PM DIBS -0.017 -0.0949 204 0905/97 03:30 PM DIBS -0.017 -0.0949 205 0905/97 12:00 PM DIBS -0.017 -0.0949 206 0905/97 12:00 PM DIBS -0.017 -0.0949 207 0905/97 12:00 PM DIBS -0.017 -0.0949 208 0905/97 12:00 PM DIBS -0.017 -0.0949 209 0905/97 12:00 PM DIBS -0.017 -0.0949 200 0905/97 12:00 PM DIBS -0.017 -0.0949 201 09005/97 12:00 PM DIBS -0.035 -0.09501 202 0905/97 12:00 PM DIBS -0.095 -0.09746 203 09005/97 12:00 PM DIBS -0.095 -0.09746 204 06005/97 12:00 PM DIBS -0.095 -0.09746 205 0905/97 12:00 PM DIBS -0.095 -0.09746 207 09008/97 12:00 PM DIBS -0.095 -0.09746 208 0605/97 12:00 PM DIBS -0.095 -0.09749 209 0606/97 12:00 PM DIBS -0.095 -0.09749 201 09008/97 12:00 PM DIBS -0.095 -0.09749 201 09008/97 12:00 PM DIBS -0.095 -0.09749	ILM-122	206	00/04/97	05:30 PM	DISS	0.067	0.05456									
202 0904/97 05:30 PM TREC 0.022 0.01646 204/97 05:30 PM TREC 0.022 0.01646 205 0904/97 05:30 PM TREC 0.032 0.01646 206 0904/97 05:30 PM TREC 0.0392 202 0904/97 12:00 AM DISS 0.045 0.0713 202 0904/97 12:00 AM DISS 0.096 0.10491 203 0906/97 03:30 PM TREC 0.374 204 0906/97 03:30 PM DISS 0.074 0.10728 204 0906/97 03:30 PM DISS 0.074 0.10728 204 0906/97 03:30 PM DISS 0.074 0.09447 204 0906/97 03:30 PM DISS 0.074 0.09499 204 0906/97 03:30 PM DISS 0.097 0.0949 204 0906/97 03:30 PM DISS 0.097 0.0949 204 0906/97 03:30 PM DISS 0.097 0.09499 204 0906/97 12:00 PM DISS 0.096 0.09786 204 0906/97 12:00 PM DISS 0.036 0.09786	ILM-123	206	09/04/97	05:30 PM	DISE	-0.022	<0.001									
202 09/04/97 05:30 PM TREC 0.022 0.01646 204 09/04/97 05:30 PM TREC 0.032 0.07113 202 08/04/97 05:30 PM TREC 0.392 202 08/04/97 12:00 AM DISS 0.045 0.0713 202 08/04/97 12:00 AM DISS 0.086 0.10491 202 08/04/97 12:00 AM DISS 0.086 0.10491 202 08/04/97 12:00 AM DISS 0.096 0.10491 202 08/04/97 12:00 AM DISS 0.097 0.09738 202 08/05/97 03:30 PM DISS 0.074 0.10728 203 08/05/97 03:30 PM DISS 0.074 0.10728 204 08/05/97 03:30 PM DISS 0.097 0.08991 204 08/05/97 03:30 PM DISS 0.097 0.08991 205 08/05/97 03:30 PM DISS 0.097 0.08991 206 08/05/97 03:30 PM DISS 0.097 0.08991 207 08/05/97 03:30 PM DISS 0.097 0.08991 208 08/05/97 03:30 PM DISS 0.097 0.08991 209 08/05/97 03:30 PM DISS 0.097 0.08991 200 08/05/97 12:00 PM DISS 0.095 0.0940 200 08/05/97 12:00 PM DISS 0.095 0.0940 201 08/05/97 12:00 PM DISS 0.095 0.09766 202 08/05/97 12:00 PM DISS 0.095 0.09768 203 08/05/97 12:00 PM DISS 0.095 0.09768 204 08/05/97 12:00 PM DISS 0.095 0.09768 207 08/05/97 12:00 PM DISS 0.095 0.09768 208 08/05/97 12:00 PM DISS 0.095 0.09769 209 08/05/97 12:00 PM DISS 0.095 0.09769 201 08/05/97 12:00 PM DISS 0.095 0.09769 202 08/05/97 12:00 PM DISS 0.095 0.09769 203 08/05/97 12:00 PM DISS 0.095 0.09769 204 08/05/97 12:00 PM DISS 0.095 0.09769 207 08/05/97 12:00 PM DISS 0.095 0.09769 208 08/05/97 12:00 PM DISS 0.095 0.09769 209 08/05/97 12:00 PM DISS 0.095 0.09769	ILM-124A	202	09/04/97	05:30 PM	DISS	900.0	0.01603			<0.024						
204 09/04/97 06:30 PM TREC 0.392 202 09/04/97 12:00 AM DISS 0.045 0.07113	ILM-124B	202	09/04/97	05:30 PM	TREC	0.022	0.01846			0.027						
202 09/04/97 05:30 PM TREC 0.392 202 09/04/97 12:00 AM DISS 0.186 202 09/04/97 12:00 AM DISS 0.077 0.09582 202 09/04/97 12:00 AM DISS 0.096 0.10491 202 09/04/97 12:00 AM DISS 0.097 0.09738 202 09/05/97 12:00 AM DISS 0.097 0.09411 204 09/05/97 03:30 PM DISS 0.074 0.10728 205 09/05/97 03:30 PM DISS 0.074 0.10728 206 09/05/97 03:30 PM DISS 0.097 0.09947 207 09/05/97 03:30 PM DISS 0.074 0.00447 208 09/05/97 03:30 PM DISS 0.097 0.09991 209 09/05/97 03:30 PM DISS 0.097 0.09991 209 09/05/97 12:00 PM DISS 0.095 0.09766 209 06/05/97 12:00 PM DISS 0.095 0.09766 209 06/05/97 12:00 PM DISS 0.095 0.09766 209 06/05/97 12:00 PM DISS 0.095 0.09768	ILM-125	204	09/04/97	05:30 PM	DISS	0.045	0.07113			<0.024						
202 09/04/97 12:00 AM DISS 0.186 202 U9/04/97 12:00 AM DISS 0.077 0.09562 202 U9/04/97 12:00 AM DISS 0.096 0.10491 202 U9/04/97 12:00 AM DISS 0.096 0.10491 202 U9/05/97 12:00 AM DISS 0.072 0.09471 204 09/05/97 03:30 PM DISS 0.075 0.0947 205 09/05/97 03:30 PM DISS 0.075 0.0947 206 09/05/97 03:30 PM DISS 0.075 0.0940 207 09/05/97 03:30 PM DISS 0.077 0.09447 208 09/05/97 03:30 PM DISS 0.077 0.09447 209 09/05/97 03:30 PM DISS 0.077 0.0949 200 09/05/97 03:30 PM DISS 0.077 0.0949 201 09/05/97 07:00 AM DISS 0.095 0.0940 202 09/05/97 12:00 PM DISS 0.095 0.0970 204 09/05/97 12:00 PM DISS 0.095 0.0970 205 09/05/97 12:00 PM DISS 0.095 0.0970 206 09/05/97 12:00 PM DISS 0.095 0.0970 207 09/05/97 12:00 PM DISS 0.095 0.0970 208 09/05/97 12:00 PM DISS 0.095 0.0970 209 09/05/97 12:00 PM DISS 0.095 0.0970 200 09/05/97 12:00 PM DISS 0.095 0.0970 201 09/05/97 12:00 PM DISS 0.095 0.0970	ILM-126A	202	09/04/97	05,30 PM	TREC	0.392				2776						
202 U9V04/97 12:00 AM DISS 0.077 0.09582 202 U9V04/97 12:00 AM DISS 0.096 0.10491 202 U9V04/97 12:00 AM DISS 0.096 0.10491 202 U9V04/97 12:00 AM DISS 0.072 0.09411 203 U9V05/97 03:00 AM DISS 0.072 0.08376 204 U9V05/97 03:30 PM DISS 0.074 0.10728 205 U9V05/97 03:30 PM DISS 0.074 0.10728 206 U9V05/97 03:30 PM DISS 0.074 0.09447 207 U9V05/97 03:30 PM DISS 0.074 0.09447 208 U9V05/97 03:30 PM DISS 0.074 0.09447 209 U9V05/97 03:30 PM DISS 0.097 0.0949 200 U9V05/97 03:30 PM DISS 0.095 0.0940 201 U9V05/97 07:00 AM DISS 0.095 0.07455 202 U9V05/97 12:00 PM DISS 0.095 0.07455 203 U5V05/97 12:00 PM DISS 0.095 0.09786 204 U5V05/97 12:00 PM DISS 0.095 0.09786 205 U5V05/97 12:00 PM DISS 0.095 0.09786 207 U5V05/97 12:00 PM DISS 0.095 0.09786 208 U5V05/97 12:00 PM DISS 0.095 0.097886 209 U5V05/97 12:00 PM DISS 0.095 0.097886 201 U5V05/97 12:00 PM DISS 0.095 0.097886 203 U5V05/97 12:00 PM DISS 0.095 0.097886 204 U5V05/97 12:00 PM DISS 0.095 0.097886 205 U5V05/97 12:00 PM DISS 0.095 0.097886 207 U5V05/97 12:00 PM DISS 0.095 0.097888	ILM-1268	202	09/04/97	12:00 AM	DISS	0.186										
202 UBV04/9F 12:00 AM DISS 0.096 0.10491 202 UBV04/9F 12:00 AM DISS 0.082 0.09738 202 UBV04/9F 12:00 AM DISS 0.072 0.09411 202 09/05/9F 03:00 AM DISS 0.075 0.08376 203 09/05/9F 03:30 AM DISS 0.075 0.0847 204 09/05/9F 03:30 PM TREC 0.374 204 09/05/9F 03:30 PM TREC 0.374 A 202 09/05/9F 03:30 PM TREC A 203 08/05/9F 0.330 PM TREC 0.374 A 204 09/05/9F 0.330 PM DISS 0.004 0.0044 B 204 06/05/9F 0.330 PM DISS 0.004 0.0044 A 202 06/05/9F 0.700 AM DISS 0.004 0.0044 B 204 06/05/9F 12:00 PM DISS <	ILM-125C	202	09/04/97	12:00 AM	DISS	0.077	0.09582									
202 09/04/97 12:00 AM DISS 0.0092 0.09/38 202 09/05/97 03:00 AM DISS 0.0072 0.09411 202 09/05/97 03:00 AM DISS 0.075 0.09411 203 09/05/97 03:30 PM TREC 0.374 203 09/05/97 03:30 PM DISS 0.074 0.10728 204 09/05/97 03:30 PM DISS 0.074 0.10728 205 09/05/97 03:30 PM DISS 0.074 0.09447 204 09/05/97 03:30 PM DISS 0.094 0.09447 204 09/05/97 03:30 PM DISS 0.097 0.0949 A 202 09/05/97 07:00 AM DISS 0.097 0.0949 B 203 09/05/97 07:00 AM DISS 0.095 0.07455 A 202 09/05/97 12:00 PM DISS 0.096 0.09786 B 203 09/05/97 12:00 PM DISS 0.096 0.09786 B 204 09/05/97 12:00 PM DISS 0.096 0.09786	ILM-126D	202	09/04/97	12:00 AM	DISS	0.096	0.10491									
202 09/04/97 12:00 AM DISS 0.072 0.09411 202 09/05/97 03:00 AM DISS 0.075 0.07863 202 09/05/97 03:30 PM TREC 0.374 202 09/05/97 03:30 PM TREC 0.374 203 09/05/97 03:30 PM DISS 0.074 0.10728 204 09/05/97 03:30 PM DISS 0.074 0.0947 204 09/05/97 03:30 PM DISS 0.007 0.0849 A 202 05/05/97 03:30 PM DISS 0.007 0.0949 B 203 06/05/97 07:20 AM DISS 0.007 0.0949 B 203 06/05/97 07:20 AM DISS 0.005 0.0745 B 203 06/05/97 12:200 PM DISS 0.005 0.09786 B 203 06/05/97 12:200 PM DISS 0.005 0.0976 201 06/05/97	II M-126E	202	09/04/97	12:00 AM	DISS	0.082	0.09738									
204 09/05/97 03:00 AM DISS 0.075 0.08376 204 09/05/97 03:00 AM DISS 0.074 0.07863 202 09/05/97 03:30 PM TREC 0.374 203 09/05/97 03:30 PM DISS 0.074 0.10728 204 09/05/97 03:30 PM DISS 0.007 0.08691 A 204 09/05/97 03:30 PM DISS 0.007 0.08691 A 202 09/05/97 03:30 PM DISS 0.007 0.08691 A 202 09/05/97 07:00 AM DISS 0.007 0.0849 B 203 06/05/97 07:00 AM DISS 0.005 0.07450 A 202 06/05/97 12:00 PM DISS 0.005 0.09760 B 203 06/05/97 12:00 PM DISS 0.005 0.09760 A 204 06/05/97 12:00 PM DISS 0.034 0.09501	II M-126	202	09/04/97	12:00 AM	DISS	0.072	0.09411									
204 09/05/97 03:30 AM PISS 0.075 0.07863 202 09/05/97 03:30 PM TREC 0.374 203 09/05/97 03:30 PM DISS 0.074 0.10728 204 09/05/97 03:30 PM DISS 0.097 0.0849 205 09/05/97 07:30 AM DISS 0.097 0.0849 205 09/05/97 07:30 AM DISS 0.096 0.0949 206 09/05/97 12:30 PM DISS 0.096 0.09786 207 09/05/97 12:30 PM DISS 0.096 0.09786 208 09/05/97 12:30 PM DISS 0.096 0.09501 209 09/05/97 12:30 PM DISS 0.0315 201 09/08/97 12:30 PM DISS 0.315	ILM-128A	2012	76/50/60	03:00 AM		0.079	0.08376									
202 09/05/97 03:30 PM TREC 0.374 202 09/05/97 03:30 PM DISS 0.074 0.10728 203 09/05/97 03:30 PM DISS 0.097 0.08447 204 09/05/97 03:30 PM DISS 0.007 0.08691 204 09/05/97 03:30 PM DISS 0.007 0.08691 204 09/05/97 07:00 AM DISS 0.004 0.0849 203 06/05/97 07:00 AM DISS 0.094 0.0849 204 06/05/97 07:00 AM DISS 0.072 0.07455 204 06/05/97 12:00 PM DISS 0.095 0.09786 204 06/05/97 12:00 PM DISS 0.084 0.09501 204 06/05/97 12:00 PM DISS 0.0316 201 06/05/97 12:00 PM DISS 0.0316 201 06/05/97 12:00 PM TREC	ILM-129	204	76/50/50	03-00 AM	DISS	0.075	0.07863									
202 09/05/97 03:30 PM DISS 0.074 0.10728 203 09/05/97 03:30 PM DISS 0.077 0.09447 204 09/05/97 03:30 PM DISS 0.097 0.08091 204 09/05/97 03:30 PM DISS 0.097 0.08091 202 09/05/97 07:30 PM DISS 0.097 0.0849 203 09/05/97 07:30 PM DISS 0.095 0.07455 204 09/05/97 12:30 PM DISS 0.095 0.07455 205 09/05/97 12:30 PM DISS 0.095 0.09788 206 09/05/97 12:30 PM DISS 0.095 0.09788 207 09/05/97 12:30 PM DISS 0.095 0.09788 208 09/05/97 12:30 PM DISS 0.095 0.09788 209 09/05/97 12:30 PM DISS 0.095 0.09788 201 09/06/97 12:30 PM DISS 0.0315 201 09/06/97 01:01 PM DISS 0.315 201 09/06/97 01:01 PM DISS 0.035 0.315 201 09/06/97 01:01 PM DISS 0.035 0.095 0.315 201 09/06/97 01:01 PM DISS 0.036 0.02409	ILM-130	202	26/90/60	03:30 PM	TREC	0.374				2454						
203 09/05/97 03:30 PM DISS 0.071 0.09447 204 09/05/97 03:30 PM DISS 0.097 0.08091 202 09/05/97 07:30 PM DISS 0.097 0.0849 203 09/05/97 07:30 AM DISS 0.084 0.0849 204 05/05/97 07:30 AM DISS 0.084 0.0849 205 09/05/97 07:30 AM DISS 0.095 0.07455 206 09/05/97 12:30 PM DISS 0.095 0.09786 207 09/05/97 12:30 PM DISS 0.095 0.09786 208 09/05/97 12:30 PM DISS 0.095 0.09786 209 09/05/97 12:30 PM DISS 0.095 0.09786 201 09/06/97 12:30 PM DISS 0.315 201 09/06/97 01:01 PM DISS 0.315 201 09/06/97 01:01 PM DISS 0.035 0.315 201 09/06/97 01:02 PM DISS 0.036 0.02409	ILM-130A	202	76/50/20	03:30 PM	DISS	0.074	0.10728			<0.024						
204 09/05/97 03:30 PM DISS 0.097 0.08991 204 09/05/97 03:30 PM DISS -0.017 <0.001	ILM-131	203	76/50/90	03:30 PM	DISS	0.071	0.09447									
204 09/05/97 03:30 PM DISS -0.017 <0.001 202 09/05/97 07:30 AM DIS3 0.084 0.0849 203 06/05/97 07:30 AM DIS3 0.095 0.07070 204 06/05/97 07:30 AM DISS 0.072 0.07455 205 06/05/97 12:30 PM DISS 0.095 0.09786 206 06/05/97 12:30 PM DISS 0.095 0.09786 207 06/05/97 12:30 PM DISS 0.095 0.09786 208 06/05/97 12:30 PM DISS 0.084 0.09501 201 06/05/97 12:30 PM DISS 0.315 201 09/08/97 01:01 PM DISS 0.315 201 09/08/97 01:01 PM DISS 0.075 0.075 0.075 3.3	ILM-131A	204	70/20/20	03:30 PM	DISS	0.097	0.08091									
202 09/05/97 07:00 AM DIS3 0.084 0.0849 203 06/05/97 07:00 AM DIS3 0.095 0.07070 204 06/05/97 07:00 AM DIS3 0.072 0.07455 202 06/05/97 12:00 PM DISS 0.106 203 06/05/97 12:00 PM DISS 0.095 0.09786 204 06/05/97 12:00 PM DISS 0.084 0.09786 204 06/05/97 12:00 PM DISS 0.084 0.09501 201 06/05/97 12:00 PM DISS 0.315 409/05/97 12:00 AM TREC: 109/05/97 12:00 AM TREC: 201 09/06/97 01:01 PM DISS 0.075 0.075 0.07533 201 09/06/97 01:02 PM DISS 0.036 0.02409	ILM-132	204	05/05/97	03:30 PM	DISS	-0.017	<0.001									
203 06/05/97 07:00 AM DIS3 0.095 0.07070 204 06/05/97 07:00 AM DIS3 0.072 0.07455 202 06/05/97 12:00 PM DISS 0.106 203 06/05/97 12:00 PM DISS 0.084 0.09786 204 06/05/97 12:00 PM DISS 0.084 0.09501 201 06/05/97 12:00 PM DISS 0.315 US/05/97 12:00 PM DISS 0.315 US/05/97 12:00 PM DISS 0.075 0.09501 201 09/08/97 12:00 PM DISS 0.315 201 09/08/97 01:01 PM DISS 0.075 0.075 0.07533 201 09/08/97 01:02 PM DISS 0.036 0.02409	ILM-1120A	202	76/20/20	MA 00:20	DISS	0.084	0.0849									
204 09/05/97 07:00 AM DISS 0.072 0.07455 202 09/05/97 12:00 PM DISS 0.095 0.09786 203 09/05/97 12:00 PM DISS 0.095 0.09786 204 09/05/97 12:00 PM DISS 0.094 0.09501 201 09/05/97 12:00 PM DISS 0.315 US/US/97 12:00 PM DISS 0.315 US/US/97 12:00 PM DISS 0.07533 201 09/08/97 01:01 PM DISS 0.075 0.075 0.075 0.07533 201 09/08/97 01:02 PM DISS 0.036 0.02409	ILM-1128B	203	76/20/30	MA 00:70	DIS3	0.095	0.07878									
202 05/05/97 12:00 PM DISS 0.108 203 06/05/97 12:00 PM DISS 0.095 0.09786 204 06/05/97 12:00 PM DISS 0.084 0.09501 201 06/05/97 12:00 PM DISS 0.315 US/US/97 12:00 PM DISS 0.315 US/US/97 12:00 PM DISS 0.054 201 09/08/97 12:00 PM DISS 0.075:33 201 09/08/97 01:01 PM DISS 0.075:33 201 09/08/97 01:02 PM DISS 0.036 0.02409	ILM-1129	204	05/05/97	MA 00.70	DISS	0.072	0.07455									
203 09/05/97 12:30 PM DISS 0.095 0.09786 204 05/05/97 12:30 PM DISS 0.084 0.09501 201 05/05/97 12:30 PM DISS 0.315 US/05/97 12:30 PM DISS 0.315 US/05/97 12:30 AM TREC: pitation Process Demonstration (Mineral Hill Water) 201 09/08/97 01:01 PM DISS 0.075:33 201 09/08/97 01:02 PM DISS 0.036 0.02409	ILM-2128A	202	76/20/50	12.00 PM	DISS	0.108										
204 05/05/97 12:00 PM DISS 0.084 0.09501 201 05/05/97 12:00 PM DISS 0.315 US/05/97 12:00 AM TREC: pitation Process Demonstration (Mineral Hill Water) 201 09/08/97 01:01 PM DISS 0.075:33 201 09/08/97 01:02 PM DISS 0.036 0.02409	ILM-2128B	203	76/00/30	12:00 PM	DISS	0.095	0.09786									
201 09/05/97 12:00 PM DISS 0.315 US/05/97 12:00 AM TREC: pitation Process Demonstration (Mineral Hill Water) 201 09/08/97 01:01 PM DISS 0.075:33 201 09/08/97 01:02 PM DISS 0.036 0.02409	ILM-2129	204	05/05/97	12:00 PM	DISS	0.084	0.09501									
DS05/97 12:00 AM TREC: pitation Process Demonstration (Mineral Hill Water) 201 09/08/97 01:01 PM DISS 0.075 0.02409	ILM-2130	201	05/02/97	12:00 PM	DISS	0.315										
201 09/08/97 01:02 PM DISS 0.035 201 09/08/97 01:01 PM DISS 0.075 201 09/08/97 01:02 PM DISS 0.036	Fecta SOL	Z	78/90/80	12:00 AM						2087.2						
201 09/08/97 01:01PM DISS 0.075 201 09/08/97 01:02 PM DISS 0.036	Iron Copre	cipitation P	rocess Dan	nonstration	(Mineral	Hill Water)										
201 09/08/97 01:02 PM DISS 0.036	BTFE 101A		09/08/97	01:01 PM		0.075	0.07533									
	BTFE 101B		26/80/60	01:02 PM		0.036	0.02409									

Sample Time As As As As Time Time Port Sampled Sampled Time Ti													Arsenic	Arsenic Speciation Results	Results	
Sample Date Time Sample (ICP) (AA) Cd Cu Fe PD Zn As (H) 1014 201 960/9647 02.00 PM 178 0.00571 (mgL)						As	As							Lotal		
Port Sampled Sampled Type (mg/L) (mg/L) <th>FIELD</th> <th>Sample</th> <th>Date</th> <th>Time</th> <th>Sample</th> <th>(ICP)</th> <th>(AA)</th> <th>8</th> <th>ವ</th> <th>Fe</th> <th>Pp</th> <th>۵</th> <th>UZ</th> <th>ΑS</th> <th>As (III)</th> <th>As (V)</th>	FIELD	Sample	Date	Time	Sample	(ICP)	(AA)	8	ವ	Fe	Pp	۵	UZ	ΑS	As (III)	As (V)
201 09068/97 02:00 PM DISS -0.003 0.00571 202 09068/97 02:00 PM TM 0.342 202 09068/97 01:01 PM DISS 0.026 0.03288 202 09068/97 01:02 PM DISS 0.004 0.00536 202 09068/97 01:05 PM DISS -0.044 0.00536 202 09068/97 01:05 PM DISS -0.044 0.00536 202 09068/97 01:05 PM DISS -0.044 0.00536 203 09068/97 01:05 PM DISS -0.043 0.0064 203 09068/97 01:05 PM DISS -0.043 0.0054 203 09068/97 01:05 PM DISS -0.022 0.0064 203 09068/97 01:05 PM DISS -0.024 0.0054 204 09068/97 01:05 PM DISS 0.028 0.00141 205 09068/97 01:05 PM DISS 0.028 </th <th>01</th> <th>Port</th> <th>Sampled</th> <th>Sampled</th> <th>Type</th> <th>(mg/L)</th> <th>(mg/L)</th> <th>(mg/L)</th> <th>(mg/L)</th> <th>(mg/L)</th> <th>(ng/L)</th> <th>(mg/L)</th> <th>(mg/L)</th> <th>(ng/L)</th> <th>(ug/L)</th> <th>(ng/L)</th>	01	Port	Sampled	Sampled	Type	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(mg/L)	(ng/L)	(mg/L)	(mg/L)	(ng/L)	(ug/L)	(ng/L)
101F 201 09/08/97 02:00 PM TM 0.342 102A 202 09/08/97 01:01 PM DISS 0.026 0.03288 102D 202 09/08/97 01:05 PM DISS 0.004 0.0042 102D 202 09/08/97 01:05 PM DISS 0.004 0.00536 102D 202 09/08/97 01:15 PM DISS 0.004 0.00536 102A 202 09/08/97 01:15 PM DISS 0.004 0.00536 102B 202 09/08/97 01:15 PM DISS 0.004 0.00536 103C 203 09/08/97 01:15 PM DISS 0.004 0.00536 103D 203 09/08/97 01:15 PM DISS 0.004 0.0054 103D 203 09/08/97 01:15 PM DISS 0.024 0.0054 103D 203 09/08/97 01:15 PM DISS 0.0044 0.0054 201D 201 <td>BTFE 101E</td> <td>201</td> <td>76/80/60</td> <td>02:00 PM</td> <td>DISS</td> <td>-0.003</td> <td>0.00571</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td>	BTFE 101E	201	76/80/60	02:00 PM	DISS	-0.003	0.00571									
102A 202 09/08/97 01:01 PM DISS 0.026 0.03288 102B 202 09/08/97 01:02 PM DISS 0.007 0.01403 102D 202 09/08/97 01:05 PM DISS -0.04 0.00336 102B 202 09/08/97 01:05 PM DISS -0.05 0.00465 102F 202 09/08/97 02:00 PM TM 0.339 0.0046 103A 203 09/08/97 01:01 PM DISS -0.04 0.0054 103B 203 09/08/97 01:05 PM DISS -0.043 0.0054 103B 203 09/08/97 01:05 PM DISS -0.024 0.0054 103B 203 09/08/97 01:05 PM DISS -0.024 0.0054 103B 203 09/08/97 01:05 PM DISS 0.028 0.0041 201B 203 09/08/97 02:05 PM DISS 0.028 0.004 201B </td <td>BTFE 101F</td> <td>201</td> <td>76/80/60</td> <td>02:00 PM</td> <td>TM</td> <td>0.342</td> <td></td> <td></td> <td></td> <td>6.269</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td>	BTFE 101F	201	76/80/60	02:00 PM	TM	0.342				6.269						
102B 202 0908/97 01:02 PM DISS 0.004 0.01403 102D 202 0908/97 01:05 PM DISS -0.013 0.00942 102D 202 0908/97 01:15 PM DISS -0.044 0.00536 102F 202 0908/97 01:15 PM DISS -0.044 0.00564 103A 202 0908/97 02:00 PM T/M 0.339 -0.004 0.00564 103B 203 0908/97 01:05 PM DISS -0.004 0.00564 103C 203 0908/97 01:05 PM DISS -0.004 0.00564 103B 203 0908/97 01:05 PM DISS -0.004 0.00564 103C 203 0908/97 01:05 PM DISS -0.004 0.00564 201B 201 0908/97 01:15 PM DISS 0.0024 0.0054 201B 201 0908/97 01:15 PM DISS 0.002 0.0024	BTFE 102A	202	76/80/60	01:01 PM	DISS	0.026	0.03288									
102C 202 09/08/97 01:05 PM DISS -0.013 0.00942 102D 202 09/08/97 01:15 PM DISS -0.04 0.00536 102F 202 09/08/97 01:15 PM DISS -0.04 0.00564 103A 202 09/08/97 02:00 PM TM 0.339 0.00167 103B 203 09/08/97 01:01 PM DISS -0.002 0.00564 103B 203 09/08/97 01:02 PM DISS -0.003 0.00564 103B 203 09/08/97 01:05 PM DISS -0.003 0.00564 103B 203 09/08/97 01:05 PM DISS -0.024 0.00564 201 09/08/97 01:05 PM DISS -0.024 0.00544 201C 09/08/97 01:15 PM DISS 0.0024 0.00544 201C 09/08/97 03:15 PM DISS 0.022 0.0044 201C 09/08/97 03:15 PM <td>BTFE 102B</td> <td>202</td> <td>09/08/97</td> <td>01:02 PM</td> <td>DISS</td> <td>0.007</td> <td>0.01403</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td>	BTFE 102B	202	09/08/97	01:02 PM	DISS	0.007	0.01403									
102D 202 09/08/97 01:15 PM DISS -0.044 0.00536 102F 202 09/08/97 02:00 PM DISS -0.05 0.00165 103A 202 09/08/97 02:00 PM TM 0.339 -0.05 103B 203 09/08/97 07:01 PM DISS -0.043 0.00504 103C 203 09/08/97 01:05 PM DISS -0.024 0.00504 103D 203 09/08/97 01:05 PM DISS -0.024 0.00504 103B 203 09/08/97 01:05 PM DISS -0.024 0.00504 103F 203 09/08/97 01:05 PM DISS 0.024 0.00504 2010 203 09/08/97 03:15 PM DISS 0.028 0.0054 2014 201 09/08/97 03:15 PM DISS 0.028 0.0054 2014 201 09/08/97 03:15 PM DISS 0.0428 0.0054	BTFE 102C	202	76/80/60	01:05 PM	DISS	-0.013	0.00942									
102E 202 09/08/97 02:00 PM DISS -0.05 0.00165 103A 202 09/08/97 02:00 PM TM 0.339 -0.005 103B 203 09/08/97 01:01 PM DISS -0.003 0.00504 103C 203 09/08/97 01:02 PM DISS -0.043 0.00504 103D 203 09/08/97 01:05 PM DISS -0.043 0.00504 103B 203 09/08/97 01:15 PM DISS -0.024 0.00572 103E 203 09/08/97 01:15 PM DISS -0.028 0.00141 201A 201B 09/08/97 01:15 PM DISS 0.028 0.00141 201B 201 09/08/97 03:15 PM DISS 0.028 0.00141 201B 201 09/08/97 03:15 PM DISS 0.028 0.00141 201B 201 09/08/97 03:15 PM DISS 0.028 0.00141	BTFE 102D	202	76/80/60	01:15 PM	DISS	-0.044	0.00536									
102F 202 09/08/97 02:00 PM TM 0.339 103A 203 09/08/97 01:01 PM DISS -0.001 0.01027 103B 203 09/08/97 01:02 PM DISS -0.003 0.00504 103D 203 09/08/97 01:05 PM DISS -0.0043 0.00504 103B 203 09/08/97 01:15 PM DISS -0.0043 0.00504 103B 203 09/08/97 01:15 PM DISS -0.024 0.00272 2014 203 09/08/97 02:00 PM TM -0.028 0.00441 201B 204 09/08/97 03:16 PM DISS 0.208 0.00441 201B 204 09/08/97 03:16 PM DISS 0.208 0.0044 201B 204 09/08/97 03:16 PM DISS 0.008 0.0409 201B 204 09/08/97 03:16 PM DISS 0.008 0.0409 202B 2	BTFE 102E	202	76/80/60	02:00 PM	DISS	-0.05	0.00165									
103A 203 09/08/97 01:01 PM DISS -0.001 0.01027 103B 203 09/08/97 01:02 PM DISS -0.032 0.00504 103C 203 09/08/97 01:05 PM DISS -0.043 0.0054 103B 203 09/08/97 01:15 PM DISS -0.024 0.00572 103F 203 09/08/97 02:00 PM TM -0.028 0.00441 2014 201 09/08/97 02:15 PM DISS 0.203 2015 204/08/97 03:15 PM DISS 0.208 2016 204/08/97 03:15 PM DISS 0.028 2017 204/08/97 03:15 PM DISS 0.028 2016 204/08/97 03:15 PM DISS 0.048 2022 09/08/97 03:15 PM DISS 0.048 2022 09/08/97 03:15 PM DISS 0.044 0.04351 2022 09/08/97 03:15 PM DISS	BTFE 102F	202	76/80/60	02:00 PM	MT	0.339				3.048						
103B 203 09/08/97 01:02 PM DISS -0.032 0.00504 103C 203 09/08/97 01:05 PM DISS -0.043 0.0054 103B 203 09/08/97 01:05 PM DISS -0.024 0.00572 103F 203 09/08/97 02:00 PM DISS 0.0223 0.00441 201A 201 09/08/97 02:00 PM TM -0.028 0.00141 201B 201 09/08/97 02:00 PM TM -0.028 0.00141 201B 201 09/08/97 03:16 PM DISS 0.208 0.00141 201C 201 09/08/97 03:16 PM DISS 0.208 0.0046 201D 201 09/08/97 03:16 PM DISS 0.086 0.0546 202D 201 09/08/97 03:16 PM DISS 0.068 0.0479 202D 202 09/08/97 03:16 PM DISS 0.058 0.0464 20	BTFE 103A	203	76/80/60	01:01 PM	DISS	-0.001	0.01027									
103C 203 09/08/97 01:05 PM DISS -0.043 0.0054 103D 203 09/08/97 01:15 PM DISS -0.024 0.00272 103E 203 09/08/97 02:00 PM DISS -0.028 0.00141 2014 203 09/08/97 02:00 PM TM -0.028 0.00141 2014 201 09/08/97 03:16 PM DISS 0.203 2015 201 09/08/97 03:16 PM DISS 0.208 2016 201 09/08/97 03:15 PM DISS 0.048 2020 09/08/97 03:15 PM DISS 0.058 0.0546 2022 09/08/97 03:15 PM DISS 0.058 0.04079 2022 09/08/97 03:15 PM DISS 0.058 0.04079 2022 09/08/97 03:15 PM DISS 0.048 0.0415 2022 09/08/97 03:15 PM DISS 0.048 0.0445 <	BTFE 103B	203	76/80/60	01:02 PM	DISS	-0.032	0.00504									
103D 203 09/08/97 01:15 PM DISS -0.024 0.00272 103F 203 09/08/97 02:00 PM TM -0.028 0.00141 201A 203 09/08/97 02:00 PM TM -0.028 0.00141 201A 204 09/08/97 02:00 PM TM -0.028 0.00141 201B 201 09/08/97 03:15 PM DISS 0.208 0.0014 201C 201 09/08/97 03:15 PM DISS 0.208 0.0546 201F 201 09/08/97 03:15 PM DISS 0.068 0.0409 201F 201 09/08/97 03:15 PM DISS 0.068 0.0409 202D 202 09/08/97 03:15 PM DISS 0.068 0.0409 202D 202 09/08/97 03:15 PM DISS 0.068 0.0409 202D 202 09/08/97 03:15 PM DISS 0.068 0.0415 202D <td>BTFE 103C</td> <td>203</td> <td>76/80/60</td> <td>01:05 PM</td> <td>DISS</td> <td>-0.043</td> <td>0.0054</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td>	BTFE 103C	203	76/80/60	01:05 PM	DISS	-0.043	0.0054									
103E 203 09/08/97 02:00 PM DISS 0.352 103F 203 09/08/97 02:00 PM TM -0.028 0.00141 201A 201 09/08/97 03:15 PM DISS 0.223 201B 201 09/08/97 03:16 PM DISS 0.208 201C 201 09/08/97 03:16 PM DISS 0.208 201D 201 09/08/97 03:15 PM DISS 0.208 201E 201 09/08/97 03:15 PM DISS 0.088 202E 201 09/08/97 04:15 PM TM 0.428 202E 202 09/08/97 04:15 PM DISS 0.064 202E 202 09/08/97 04:15 PM DISS 0.068 0.04079 202E 202 09/08/97 04:15 PM DISS 0.068 0.0446 0.0448 202E 202 09/08/97 04:15 PM DISS 0.006 0.01546	BTFE 103D	203	76/80/60	01:15 PM	DISS	-0.024	0.00272									
103F 203 09/08/97 02:00 PM TM -0.028 0.00141 201A 201 09/08/97 03:15 PM DISS 0.208 201C 201 09/08/97 03:15 PM DISS 0.208 201C 201 09/08/97 03:15 PM DISS 0.208 201E 201 09/08/97 03:15 PM DISS 0.186 201E 201 09/08/97 03:15 PM DISS 0.068 202E 201 09/08/97 04:15 PM TM 0.428 202E 202 09/08/97 03:15 PM DISS 0.064 202E 202 09/08/97 03:15 PM DISS 0.0435 202E 202 09/08/97 03:15 PM DISS 0.068 0.04079 202E 202 09/08/97 03:15 PM DISS 0.035 0.0346 202E 202 09/08/97 03:15 PM DISS 0.005 0.01546 203 <	BTFE 103E	203	76/80/60	02:00 PM	DISS	0.352										
2014 201 09/08/97 03:15 PM DISS 0.203 2018 201 09/08/97 03:16 PM DISS 0.208 201C 201 09/08/97 03:16 PM DISS 0.186 201D 201 09/08/97 03:15 PM DISS 0.186 201E 201 09/08/97 03:15 PM DISS 0.168 201F 201 09/08/97 04:15 PM DISS 0.064 202A 202 09/08/97 03:15 PM DISS 0.064 0.04079 202C 202 09/08/97 03:15 PM DISS 0.068 0.04079 202C 202 09/08/97 03:15 PM DISS 0.035 0.034079 202E 202 09/08/97 03:15 PM DISS 0.005 0.04079 203A 203 09/08/97 03:15 PM DISS 0.005 0.014079 203B 203 09/08/97 03:15 PM DISS 0.005 0.01546	BTFE 103F	203	76/80/60	02:00 PM	TM	-0.028	0.00141			0.354						
2016 201 09/08/97 03:16 PM DISS 0.208 201C 201 09/08/97 03:05 PM DISS 0.186 201D 201 09/08/97 03:05 PM DISS 0.186 201E 201 09/08/97 03:15 PM DISS 0.068 201F 201 09/08/97 04:15 PM TM 0.428 202A 202 09/08/97 03:15 PM DISS 0.084 202D 202 09/08/97 03:15 PM DISS 0.054 202D 202 09/08/97 03:15 PM DISS 0.054 202D 202 09/08/97 03:15 PM DISS 0.007 0.02599 202E 202 09/08/97 03:15 PM DISS 0.007 0.02599 202E 202 09/08/97 03:15 PM DISS 0.007 0.02599 203E 203 09/08/97 03:15 PM DISS 0.004 0.01516 203	BTFE 201A	201	76/80/60	03:15 PM	DISS	0.223										
201C 201 09/08/97 03:05 PM DISS 0.186 201D 201 09/08/97 03:15 PM DISS 0.208 201E 201 09/08/97 04:15 PM DISS 0.168 201E 201 09/08/97 04:15 PM TM 0.048 202A 202 09/08/97 03:15 PM DISS 0.089 0.0546 202B 202 09/08/97 03:15 PM DISS 0.068 0.04079 202C 202 09/08/97 03:15 PM DISS 0.051 0.04079 202B 202 09/08/97 03:15 PM DISS 0.035 0.04079 202E 202 09/08/97 04:15 PM DISS 0.035 0.0316 202E 202 09/08/97 04:15 PM DISS 0.044 0.01833 203E 203 09/08/97 03:15 PM DISS 0.006 0.01516 203E 203 09/08/97 03:15 PM DISS <td>BTFE 201B</td> <td>201</td> <td>09/08/97</td> <td>03:16 PM</td> <td>DISS</td> <td>0.208</td> <td></td>	BTFE 201B	201	09/08/97	03:16 PM	DISS	0.208										
201D 201 09/08/97 03:15 PM DISS 0.208 201E 201 09/08/97 04:15 PM DISS 0.168 201F 201 09/08/97 04:15 PM TM 0.428 202A 202 09/08/97 03:15 PM DISS 0.089 0.0546 202C 202 09/08/97 03:15 PM DISS 0.061 0.04079 202C 202 09/08/97 03:15 PM DISS 0.058 0.04079 202E 202 09/08/97 03:15 PM DISS 0.035 0.0316 202E 202 09/08/97 04:15 PM DISS 0.035 0.0316 202E 202 09/08/97 04:15 PM TM 0.368 203E 202 09/08/97 04:15 PM DISS 0.006 0.01516 203E 203 09/08/97 03:15 PM DISS 0.006 0.01516 203E 203 09/08/97 03:15 PM DISS	BTFE 201C	201	76/80/60	03:05 PM	DISS	0.186										
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202F 202 09/08/97 04:15 PM TM 0.368 203A 203 09/08/97 03:15 PM DISS 0.044 0.01833 203B 203 09/08/97 03:15 PM DISS 0.006 0.01516 203C 203 09/08/97 03:05 PM DISS 0.039 0.01397 203E 203 09/08/97 03:15 PM DISS -0.002 0.01295 203E 203 09/08/97 04:15 PM DISS 0 0.01126 203F 203 09/08/97 04:15 PM TM 0.42	BTFE 202E	202	09/08/97	04:15 PM	DISS	0.007	0.02599									
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203 09/08/97 04:15 PM DISS 0 0.01126 203 09/08/97 04:15 PM TM 0.42	BTFE 203D	203	76/80/60	03:15 PM	DISS	-0.002	0.01295									
203 09/08/97 04:15 PM TM 0.42	BTFE 203E	203	09/08/97	04:15 PM	DISS	0	0.01126									
	BTFE 203F	203	26/80/60	04:15 PM	TM	0.42				1.579						